

3.3 Damping Effect in the Structure of Austenitic Steel After the Treatment by Radial Shear Rolling + Cold Rolling

The damping effect in the heterogeneous internal structure of austenitic steel is studied by scratch testing at the indenter loads 50, 100, and 200 mN. The damping effect is distinct at all the studied loads. Figures 14 and 15 show groove profiles at the initial and treated surfaces of the steel specimens at the indenter load 100 mN.

As can be seen from Fig. 14, the highly plastic initial steel forms a pronounced smooth groove ~170 nm in depth. Bands of plastic shear are visible at the groove surface. A qualitatively different type of grooves is formed at the specimen surface treated by radial shear rolling + cold rolling (Fig. 15). The groove depth is only 15 nm, and its central zone is extruded ~15 nm above the initial surface. No traces of

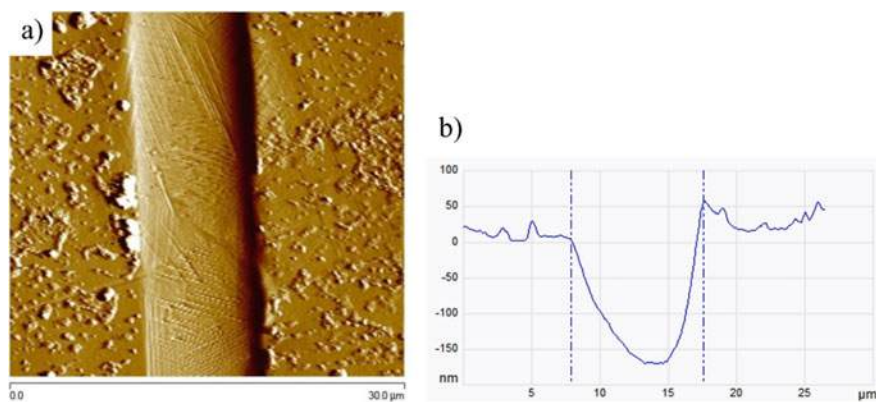


Fig. 14 Smooth surface of a deep groove (a) and its profile (b) for the steel in the initial state

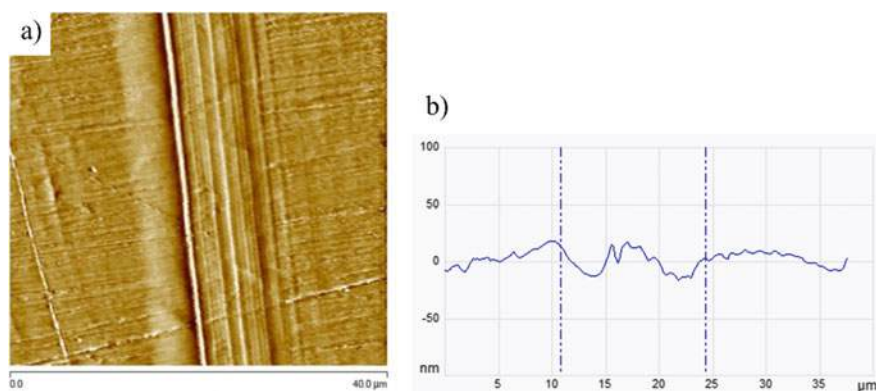


Fig. 15 Corrugated surface of a groove with the damping effect (a) and its profile (b) for the steel treated by radial shear rolling with subsequent cold rolling

Table 1 Nanohardness H , elastic modulus E^* , and shape recovery factor R at different states of the steel

State	Characteristics		
	H (GPa)	E^* (GPa)	R (%)
Initial	3.7 ± 0.2	229 ± 31	7
After radial shear and cold rolling	4.9 ± 0.3	190 ± 12	14

plastic shear are observed at the groove surface in the treated steel (Fig. 15a). Plastic extrusion is due to nanostructural transformations. The damping effect is very strong, while it is associated with a nonuniform stress distribution at the interface between the groove bottom and the substrate material.

Mechanical characteristics of the austenitic steel in different states are presented in Table 1.

From the tabulated data it is seen that complex radial shear and cold rolling increases the nanohardness H , decreases the elastic modulus E^* , and doubles the shape recovery factor R . The ductile mode of the material extrusion in scratch testing appears not only within the groove, but also in the material on its left and right. All these zones have no traces of plastic shear, but the stress distribution heterogeneity is clearly manifested geometrically at the interface between the surface layer plastically deformed by the moving indenter and the elastic substrate.

3.4 Influence of the Treatment by Radial Shear Rolling + Cold Rolling on the Development of Gigacycle Fatigue and Wear Resistance of Austenitic Steel

Nanostructuring of the austenitic steel and the strong damping effect of the material in scratch testing should increase the fatigue life of the crystal lattice [14, 15]. Moreover, it is known that nanostructuring of a material promotes the development of gigacycle fatigue. This is fully confirmed for the austenitic steel processed by radial shear rolling with subsequent smooth-roll cold rolling. The investigation results are shown in Table 2.

Table 2 Influence of the complex treatment of austenitic steel by radial shear rolling (RShR) + cold rolling (CR) on the characteristics of its fatigue life and wear resistance

Characteristics	State		
	Initial	After RShR	After RShR + CR
The number of cycles to fatigue failure	3 mln		65 mln
Wear rate coefficient, $10^5 \text{ mm}^3/\text{N m}$	8.58	8.25	1.30

Table 2 presents two fundamentally important results. First, the obtained data confirm high-cycle fatigue failure (3×10^6 cycles to failure) in the initial austenitic steel and gigacycle fatigue failure (more than 65×10^6 cycles to failure) after radial shear and cold rolling of this steel.¹ Gigacycle fatigue is usually realized at a significant reduction in external applied stresses [15, 16]; however, the tabulated results are derived at a high external stress. In other words, at high external stresses the transition from high-cycle to gigacycle fatigue can occur due to a specific internal substructure formed at the nanoscale structural level. Secondly, the wear resistance of austenitic steel does not vary after high-temperature radial shear rolling. To achieve this requires additional cold rolling that forms bcc ferrite zones in austenite grains of the steel. The mechanism of formation of wear particles is also associated with fatigue failure of the tribocontact. When the counterbody moves along the flat surface of the material, each mesovolume is first compressed and then stretched. Such cyclic deformation causes fatigue fragmentation of the material and the formation of wear particles.

A heterogeneous hierarchical structure formed in austenitic steel during radial shear and cold rolling effectively functions in tribological conditions. When the counterbody compresses mesovolumes of the heterogeneous austenitic steel, bcc ferrite grains are elastically compressed and hcp ϵ -martensite laths are embedded into the close-packed fcc austenite structure by the mechanism of forward + reverse martensitic transformation. In subsequent tension of this mesovolume, the hcp ϵ -martensite laths are recovered at the interstitial nanoscale structural states, and local stresses in the bcc ferrite grains relax. These processes are reversible and significantly retard plastic deformation, cracking, and the formation of wear particles. We emphasize that this effect is also associated with reversible structural transformations at the nanoscale structural level, where nanoscale mesoscopic structural states can exist at the lattice curvature interstices.

4 Structural Turbulence and Gigacycle-Fatigue Processes in a Solid with Lattice Curvature

4.1 *Structural Turbulence of Plastic Flow at Lattice Curvature and in the Presence of Nanoscale Mesoscopic Structural States at Its Curvature Interstices*

No turbulent plastic flow can exist in a translation-invariant crystal. However, the appearance of lattice curvature zones and of nanoscale mesoscopic structural states at the lattice curvature interstices radically changes the mechanisms of plastic deformation and fracture of solids. This concerns the effect of plastic distortion [11],

¹At the time of publication the cyclic loading experiment for specimens after radial shear and cold rolling is being continued.

formation of a vortical filamentary mesosubstructure [9, 12], structural turbulence, and dynamic rotations [2, 10, 13].

Structural turbulence of plastic flow was predicted when modeling grain-boundary sliding by the excitable cellular automaton method for grain boundaries that lack translational invariance [17]. In this case, consideration was given to lattice curvature at grain boundaries and in near-boundary regions.

Clusters of excess vacancies, that number in ~ 500 at the cluster size ~ 3.5 nm, in localized strain bands were viewed by Matsukawa and Zinkle [18] in the transmission electron microscope column in tension of a gold foil. Tetrahedra of stacking faults form in such vacancy clusters, which can move in the $\langle 110 \rangle$ direction at the migration energy $E_t = 0.19$ eV. Recall that the migration energy of a single vacancy in gold is $E_v = 0.85$ eV. This means that the migration of vacancy tetrahedra is not a diffusion process, but it is associated with structural transformations of nanostructured tetrahedra in the $\langle 110 \rangle$ direction.

As noted above, coherent scattering regions of the size ~ 40 nm appear in the structure of the martensite phase, whose volume amounts to as much as 85.6%. This is a very important nanostructural element, which contributes to the formation of nanostructured grains in austenite and ferrite grains, as demonstrated above in Figs. 1 and 2. Obviously, both in localized strain bands and in nonequilibrium martensite laths in Figs. 7 and 8, a variety of structural configurations can form: highly mobile stacking fault tetrahedra, slow-moving stacking fault octahedra, misoriented nanofragments, nanograins of various composition, including nano carbides, nanocarbonitrides, and others. The presence of the 85.6% nonequilibrium martensite phase in the metastable austenitic steel with coherent scattering regions of the size ~ 40 nm makes possible reversible structural-phase transformations in the steel specimens under cyclic loading. The mechanism of such transformations is discussed below.

4.2 Influence of the Mechanism of Reversible Structural-Phase Transformations on Gigacycle Fatigue and Wear Resistance Increase in Austenitic Steel After Radial Shear and Cold Rolling

Since individual volumes of a specimen periodically undergo alternating tension-compression under cyclic loading, this process can be reversible without cracking only subject to the condition of reversible structural-phase transformations. Nanostructured fcc austenite grains and bcc ferrite zones have different yield stresses and are surrounded by the martensite phase, which arises on the basis of interstitial nanoscale structural states in lattice curvature zones that lack translational invariance. Laths of the hcp ϵ martensite in compression can be embedded into the fcc austenite structure, transforming into its close-packed configuration. This governs inelastic compression deformation of the specimen. Laths of the bcc α' martensite in compression

will elastically change the spatial orientation of covalent d bonds and generate local stresses. In tension under cyclic loading, elastic stresses in the bcc ferrite zone will relax and cause a recovery of the hcp ϵ martensite in the austenite grains, implementing inelastic tensile deformation. Such processes of structural transformations are reversible in nanostructured materials [19–21], which determines the damping effect in their structure under cyclic loading, an increase in the gigacycle fatigue life and wear resistance.

A similar damping effect is revealed during scratch testing (Fig. 15). When the indenter moves during scratch testing, the martensite phase is first compressed. Laths of the hcp ϵ martensite transform their structure into the fcc lattice of close-packed austenite. Spatially oriented along the cube diagonals, structural elements of the bcc α' martensite associated with d electrons undergo quasi-elastic compression. After the indenter passes, the ϵ -martensite recovers its hcp structure, and the groove in Fig. 15 exhibits a damping effect. Relaxation of high local elastic stresses in the α' -martensite initiates the groove recovery after the indenter passes. Thus, structural transformations in austenitic steel after high-temperature radial shear and cold rolling are indeed reversible under cyclic external influences.

An increase in the fatigue life of austenitic steel, when loaded below the yield stress of a translation-invariant material, is explained by the nonequilibrium nanostructured martensite structure produced by radial shear and cold rolling and associated with lattice curvature. An important functional role is played by the spatial distribution of ϵ - and α' -martensite laths [22]. This distribution governs a complex spatial distribution of lattice curvature, different nanoscale mesoscopic structural states, the appearance of high local internal stresses, and the possibility of gigacycle fatigue without fatigue cracking at sufficiently high external stresses.

5 Conclusions

The description of a deformable solid as a multiscale hierarchically organized system is usually limited in the literature to a microscale structural level, where strain-induced defects of a translation-invariant crystal lattice are considered. An important role in the problem of the mechanical behavior of materials is played by curvature of the crystal lattice, in whose interstices nanoscale mesoscopic structural states arise [9, 10, 22]. In the present study, such nanoscale mesoscopic structural states were formed in Fe–Cr–Mn austenitic stainless steel using the complex treatment by multistage high-temperature radial shear rolling with subsequent smooth-roll cold rolling to the resulting plastic strain 1.8–2.0.

Such complex treatment causes the formation of nanostructured fcc austenite grains in the steel, bcc ferrite zones, and lattice curvature of nonequilibrium ϵ - and α' -martensite phases in the interstitial space based on nanoscale mesoscopic structural states. Under mechanical loading, the nonequilibrium heterogeneous martensitic structure of the specimens undergoes reversible structural-phase transformations, which are responsible for a nanocrystalline structure of the material, a vortical

filamentary structure and dynamic rotations on the fracture surface, an increased wear resistance, and the transition of high-cycle fatigue life of the initial material to gigacycle fatigue without reducing external applied stress.

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Autowave Mechanics of Plastic Flow



Lev B. Zuev

Abstract The notions of plastic flow localization are reviewed here. It has been shown that each type of localized plasticity pattern corresponds to a given stage of deformation hardening. In the course of plastic flow development a changeover in the types of localization patterns occurs. The types of localization patterns are limited to a total of four pattern types. A correspondence has been set up between the emergent localization pattern and the respective flow stage. It is found that the localization patterns are manifestations of the autowave nature of plastic flow localization process, with each pattern type corresponding to a definite type of autowave. Propagation velocity, dispersion and grain size dependence of wavelength have been determined experimentally for the phase autowave. An elastic-plastic strain invariant has also been introduced to relate the elastic and plastic properties of the deforming medium. It is found that the autowave's characteristics follow directly from the latter invariant. A hypothetical quasi-particle has been introduced which correlates with the localized plasticity autowave; the probable properties of the quasi-particle have been estimated. Taking the quasi-particle approach, the characteristics of the plastic flow localization process are considered herein.

Keywords Elasticity · Plasticity · Localization · Crystal lattice · Self-organization · Autowaves · Quasi-particle

1 Introduction. General Consideration

In past few decades, the nature and salient features of plastic deformation in solids were investigated. A wealth of new experimental data has been collected, which add strong support to our understanding of plasticity problem. Naturally, we can do no more than mention a few experimental and theoretical studies related to dislocation physics and solids mechanics, e.g. [1–14]. We have recently made a significant

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discovery that the deforming medium is a self-organizing system, which is in a state far from thermodynamic equilibrium; such media are addressed by [15–24]. Moreover, the plastic flow in solids is found to have a space-time periodic nature, which is discussed at length by [25–28]. On the macro-scale level the plastic deformation exhibits an inhomogeneous localization behavior from yield point to failure. Hence, localization is a general feature of the plastic flow process, which should be taken properly into account to markedly advance our understanding of the deforming medium stratification into alternating deforming and non-deforming layers about 10^{-2} m thick. Similar layers form localized plasticity pattern.

In this line of research, considerable experimental study has been given to the problem of plastic deformation macrolocalization; the investigation results were summarized in a monograph by [29].¹ In what follows, we discuss new approaches to the same problem.

1.1 Experimental Technique

The experimental procedure was as follows. The flat samples having work part $50 \times 6 \times 2$ mm were tested in tension along the axis x at a rate of $3.5 \times 10^{-5} \text{ s}^{-1}$ in a test machine at 300 K. The non-metallic materials were subjected to compression. Traditional ‘stress–strain’, $\sigma(\varepsilon)$, diagrams recording were completed by double-exposure speckle photography [30] for reconstruction of the displacement vector field $\mathbf{r}(x, y)$. Special device for these purposes has field of vision ~ 100 mm, real-time mode of operation and spatial resolution $\sim 1 \dots 2 \text{ } \mu\text{m}$. According to [31], the plastic distortion tensor for plane stressed state is

$$\beta^{(p)} = \nabla \mathbf{r}(x, y) = \begin{bmatrix} \varepsilon_{xx} & \varepsilon_{xy} \\ \varepsilon_{yx} & \varepsilon_{yy} \end{bmatrix} + \omega_z. \quad (1)$$

Longitudinal, ε_{xx} , transverse, ε_{yy} , shear $\varepsilon_{xy} = \varepsilon_{yx}$ components and rotation ones, ω_z can be calculated for different points of the test sample.

This developed method enables visualization of the deformation inhomogeneities. Thus, Fig. 1a demonstrates a macro-photograph of the deforming material structure and $\varepsilon_{xx}(x, y)$ distribution for this case. The diagram $X(t)$ is shown in Fig. 1b (here X is the x -coordinate of deformation nucleus and t is time); it illustrates the procedure employed for measuring the spatial and temporal periods of the deformation process, i.e. the values λ , T and rate of the strain nucleus, $V_{aw} = \lambda/T$.

¹Professor S. G. Psakhie has promoted this book as the Editor.

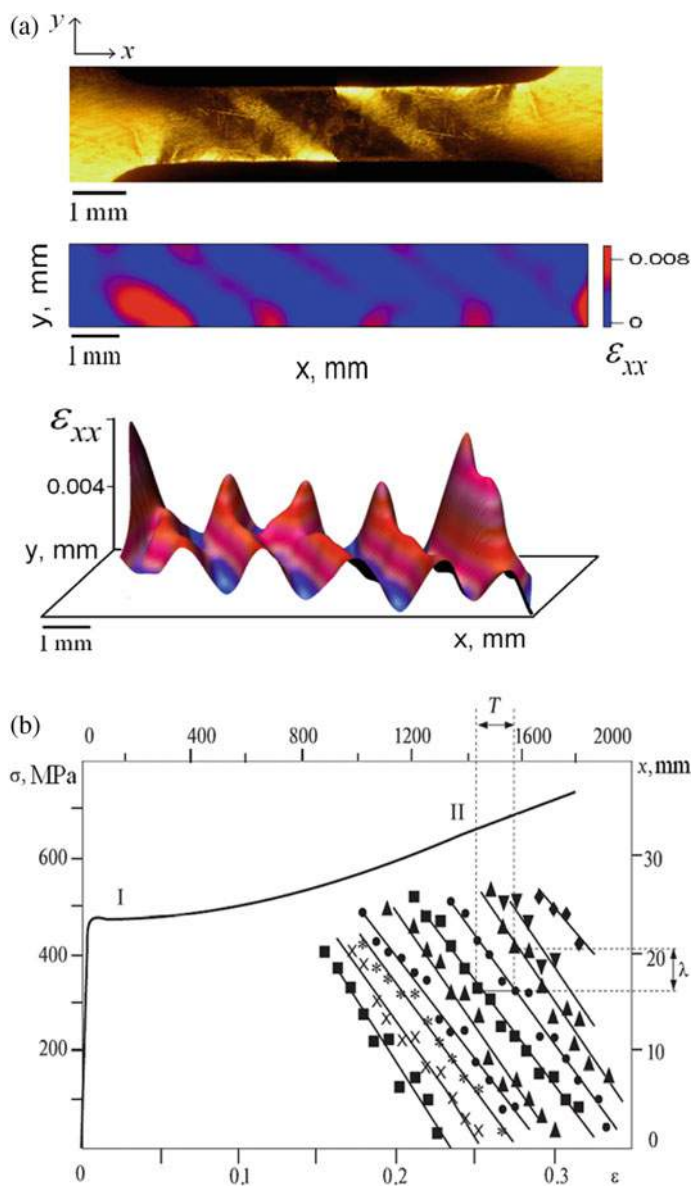


Fig. 1 Localized plastic flow pattern observed for the test sample of Fe-3 wt% Si alloy: **a** macrophotograph and distribution of strains ϵ_{xx} (half-tone photograph) **b** illustration of the method for measuring λ and T values

1.2 Studied Materials

We have studied by now nearly forty single-crystal and polycrystalline metals, alloys and other materials, which differ in chemical composition, crystal lattice type (FCC, BCC or HCP), grain size and deformation mechanism, i.e. dislocation glide [32], twinning [33] or phase transformation induced plasticity [34]. Later our study has been extended to ceramics, single alkali halide crystals and rocks.

1.3 Preliminary Results

The localization behavior of plastic deformation is its most salient feature. Thus, space-time periodic structures, so-called deformation patterns, emerge in the deforming sample from the yield limit to failure by constant-rate tensile loading. The following features are common to all the localization patterns observed thus far:

- *localization structures will occur spontaneously in the sample by constant-rate loading in the absence of any specific action from the outside;*
- *in the course of plastic deformation a changeover in the types of localized plasticity patterns is observed;*
- *due to work hardening, the deforming medium's defect structure undergoes irreversible changes, which are suggestive of its non-linearity and are reflected in the emergent patterns.*

Recent independent evidence supports the validity of the present conception about the macroscopic localization by deformation [35–37]. Acharya et al. [38] observed a localization nucleus traveling along the extension axis in the single Cu crystal at the linear work hardening stage; a stationary localization pattern at the parabolic work hardening stage in the samples of Fe-Mn alloy was described by [39].

2 Deformation Pattern. Localized Plastic Flow Viewed as Autowaves

The plastic flow has an attribute, which is common to all deforming solids. On the macro-scale level, the deformation is found to exhibit a localization behavior from the yield point to failure. In the cause of plastic flow, plastic flow curve would occur by stages; each work hardening stage involves a certain dislocation mechanism [40–42]. To gain an insight about the nature of plasticity, the existence of explicit connection between the above two attributes must be discovered.

2.1 Plastic Flow Stages and Localized Plasticity Patterns

We focused our efforts on proving this connection. To provide a proof for this assertion, a localized plasticity pattern was matched against the respective work hardening stage. These can be readily distinguished on the stress-strain curve $\sigma(\varepsilon)$, describing by the Lüdwick equation [43]

$$\sigma(\varepsilon) = \sigma_y + K\varepsilon^n, \quad (2)$$

where σ_y is the yield point and K is hardening coefficient. The convenient characteristic of the deformation process is the exponent $n = (\ln \varepsilon)^{-1} \cdot \ln[(\sigma - \sigma_y)/K]$.

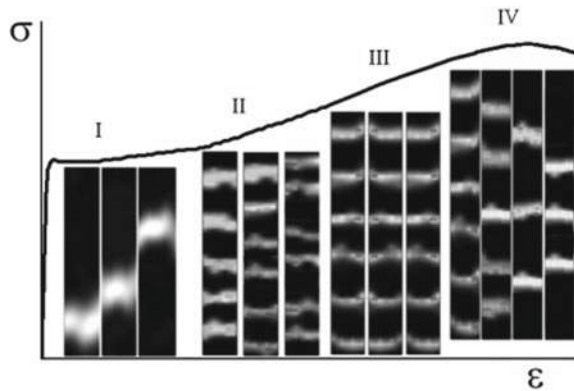
According to [43] and [44], the exponent n varies with the stages of plastic deformation process as

- for the yield plateau or easy glide stage, $n \approx 0$;
- for the linear stage of work hardening, $n = 1$;
- for the parabolic stage of work hardening, $n = 1/2$,
- for the prefailure stage, $1/2 > n > 0$.

A set of kinetic diagrams $X(t)$ was obtained simultaneously for the above stages; it is schematized in Fig. 2. Similar sets were plotted for all deforming materials, no matter what their microstructure or deformation mechanism. The localization patterns arise in a consecutive order that is governed by the work hardening law $\theta(\varepsilon)$ alone. The emergent localization patterns can be distinguished from the dependencies $X(t)$ obtained for the work hardening stages. The distinctive features of localization patterns remain the same; differences are quantitative ones. It should be pointed out that plastic deformation inhomogeneities have to be correlated for the entire material volume. The localized plasticity nuclei are distributed periodically; in all studied materials the distance between nuclei $\lambda \approx 10^{-2}$ m.

A qualitative analysis of experimental data rests on the observation that these regularities serve to provide a unified explanation of the plastic flow behavior. A

Fig. 2 Plastic flow development. Schematic representation of autowave evolution process



total of four types of localized plasticity patterns have been observed experimentally for all studied materials. A definite type of localization pattern would emerge at each flow stage. It is thus asserted that

- yield plateau is identified with a single mobile plastic flow nucleus;
- linear work hardening stage, with a set of equidistant moving nuclei;
- parabolic work hardening stage, with a set of equidistant stationary nuclei;
- prefailure stage, with a set of moving nuclei, which are forerunners of failure.

Taken together, these results provide a reliable proof that one-to-one correspondence exists between the localization patterns on the one hand, and the respective plastic flow stages on the other.

2.2 Localized Plastic Flow Autowaves

A key aspect of this many-faceted problem to be dealt with is the nature of localized plasticity. Our basic viewpoint is that there are striking parallels between the localized plasticity patterns described herein and the dissipative structures synergetics deals with. We suggest that the dissipative structures are *autowaves* [45], *self-excited waves* [46] or *pseudo-waves* [15]. Special modes of these structures, which are also known as switching autowaves, phase autowaves or stationary dissipative structures, have been studied in detail for a number of chemical and biological systems.

The difference between the wave and autowave processes needs careful explanation. The majority of well-known wave processes are described by functions $\sin(\omega t - kx)$, which are solutions of hyperbolic differential equations of the type $\ddot{Y} = c^2 Y''$. Here the value c is wave propagation rate, which is a finite quantity determined by material characteristics. The second derivative with respect to time is applicable to reversible physical processes alone, e.g. elastic deformation.

The autowaves have long been recognized as solutions to parabolic differential equations of the type $\dot{Y} = \varphi(x, y) + \kappa Y''$ [47]. These equations can be derived formally by adding to the right part of the equation $\dot{Y} = \kappa Y''$ the nonlinear function $\varphi(x, y)$; the value κ is a transport coefficient having the dimensionality $L^2 \cdot T^{-1}$. The availability of the first derivative with respect to time implies that the above equations are suitable for addressing irreversible processes similar to those involved in the plastic deformation.

The speculation that the localization patterns in question are equivalent to autowaves was originally prompted by a formal resemblance between the two kinds of phenomena. In what follows, we provide unequivocal physical evidence for the validity of this viewpoint by focusing our attention on the autowave nature of processes of interest. Thus, the well-known Lüders front can be regarded as a boundary between the elastically and plastically deforming material volumes. As the Lüders front propagates along the tensile sample, it leaves behind an ever increasing volume of deformed material [48]. Due to the structural changes, the deforming material volume acquires a new state, which is characterized by increasing density of

defects; its deformation occurs via dislocation glide mechanism. With growing total deformation, the plastic flow will exhibit an intermittent behavior on the macro-scale level. Therefore, the Lüders band propagation is regarded as a *switching autowave*. A different scenario is realized at the linear work hardening stage where a set of mobile nuclei is observed. The nuclei move at a constant rate along the test sample (see Fig. 2). In this case, the phase constancy condition is fulfilled, i.e. $\omega t - kx = \text{const.}$ This pattern will be designated a *phase autowave*. The linear stage over, the parabolic stage of work hardening begins for $n = 1/2$ and $V_{aw} = 0$ (see Fig. 2). At this stage the emergent localization pattern fits the definition of a *stationary dissipative structure* [15, 16]. At the prefailure stage the deformation development is nearing completion. For $n < 1/2$, *collapse of the autowave* would be observed [49, 50]. Thus, the types of localized plasticity patterns have been unambiguously identified with the respective modes of autowave processes.

It is therefore concluded that

- a solitary localized deformation nucleus traveling at the yield plateau is a switching autowave;
- a set of equidistant localization nuclei propagating at a constant rate along the sample at the linear work hardening stage is a phase autowave;
- a set of stationary equidistant localization nuclei emergent at Taylor's stage corresponds to a stationary dissipative structure;
- a pattern of synchronously moving nuclei, which finally merge at the prefailure stage, fits neatly the definition of collapse of localized plasticity autowave.

Taken together, these regularities are *Correspondence rule*. Accordingly, it can thus be asserted that the plastic flow process occurring in the deforming can be addressed as continuous evolution of localized plastic flow autowaves. Hence, it can be claimed with confidence that the transition from one flow stage to the next involves a changeover in the types of autowaves generated by the deformation. With due regard to the correspondence rule, it is maintained that the plastic flow stages and the respective autowaves modes are closely related. This is favorable ground for inferring that the process of localized plastic flow is evolution of autowave patterns. Due to a changeover in the flow stages, the autowaves will emerge from a random strain distribution in an orderly sequence (Fig. 2): elastic wave \rightarrow switching autowave \rightarrow phase autowave \rightarrow stationary dissipative structure \rightarrow collapse of autowave. In some materials, however, individual stages might be missing and this sequence can be broken.

It is necessary to remind here that a special-purpose reaction cell has to be designed for carrying out experimental investigations of autowaves in chemistry or biology. Such cells differ widely in type and size, depending on the kind of studied system and its chemical composition as well as the kinetics of chemical reactions involved, temperatures employed, etc. However, it is found for plastic deformation that the autowaves will be generated spontaneously in the tensile sample practically at any temperature. From this point of view, the deforming solid can be regarded as a universal reaction cell [51], which can be conveniently used for modelling and studying the generation and evolution of various autowave modes.

2.3 Autowaves Observed for the Linear Work Hardening Stage

The plastic flow exhibit generally a regular localization behavior, which is markedly pronounced at the linear work hardening stage. In this case, the localization nuclei move in a concerted manner at a constant rate along the sample, forming a phase autowave. The experimental data on propagation rate, dispersion and material structure response obtained for these autowaves are demonstrated in Fig. 3.

The propagation rates of localized plasticity autowaves in all studied materials are in the range $10^{-5} \leq V_{aw} \leq 10^{-4}$; they depend solely upon the work hardening coefficient, $\theta = E^{-1} \cdot d\sigma/d\varepsilon$, and are given as (Fig. 3a)

$$V_{aw}(\theta) = V_0 + \Xi/\theta \sim \theta^{-1}, \quad (3)$$

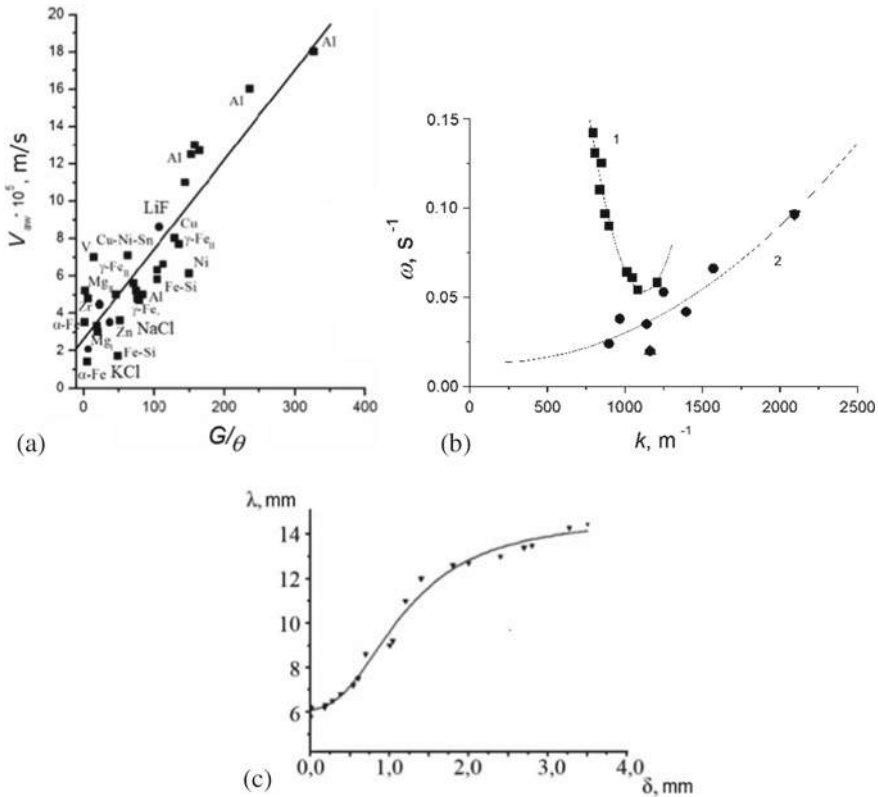


Fig. 3 Characteristics of localized plastic flow autowaves: **a** autowave rate as a function of work hardening coefficient for all studied materials; **b** dispersion observed for γ -Fe single crystals (2) and polycrystalline Al (1); **c** autowave length as a function of grain size plotted for polycrystalline Al

where $V_0 < \Xi$ are empirical constants, having the dimensionality of rate.

We have also obtained dispersion relation, $\omega(k)$ (here $\omega = 2\pi/T$ is frequency and $k = 2\pi/\lambda$ is wave number) for localized plasticity autowaves [29, 52–55]. This relation has quadratic form (Fig. 3b)

$$\omega(k) = \omega_0 + \alpha(k - k_0)^2. \quad (4)$$

Using the values $\omega = \omega_0 \cdot \tilde{\omega}$ and $k = k_0 + \frac{\tilde{k}}{\sqrt{\alpha/\omega_0}}$ (here $\tilde{\omega}$ and \tilde{k} are dimensionless frequency and wave number), Eq. (4) can reduce to the form $\tilde{\omega} = 1 + \tilde{k}^2$.

Finally, the grain size dependence of autowave length illustrated in Fig. 3c has the form of logistic curve

$$\lambda(\delta) = \lambda_0 + \frac{a_1/a_2}{1 + C \cdot \exp(-a_1\delta)}, \quad (5)$$

where a_1 , and a_2 are empirical coefficients, $\lambda_0 \approx 5$ mm and $C \approx 2.25$. The inflection point of Eq. (5) is found from the condition $d^2\lambda/d\delta^2 = 0$; this corresponds to the boundary value of grain size $\delta = \delta_0 \approx 0.15 \dots 0.2$ mm. The dependence $\lambda(\delta)$ has two limiting cases, i.e. $\lambda \sim \exp(\delta/\delta_0)$ for $\delta < \delta_0$ and $\lambda \sim \ln(\delta/\delta_0)$ for $\delta > \delta_0$. The quantity λ generally depends only weakly on the structural characteristics of the deforming medium. Thus variation in the grain size of polycrystalline Al from 5 μm to 5 mm corresponds to a 2.5-fold increase in the value λ [56].

Thus, the most significant features of localized plastic flow at the linear stage of work hardening are specified herein by Eqs. (3) through (5). By way of summing up our findings, we contend that in the course of plastic flow a large-scale deformation structure would form. Its elements are characterized by the nontrivial dependence $V_{aw}(\theta) \sim \theta^{-1}$, the quadratic dispersion law $\tilde{\omega} = 1 + \tilde{k}^2$ and the logistic dependence of autowave length on material structure, $\lambda(\delta)$.

2.4 Plastic Flow Viewed as Self-organization of the Deforming Medium

We hypothesize herein that the localization of plastic flow might be regarded as a process of self-organization occurring spontaneously in an open thermodynamic system. The validity of our hypothesis can be objectively confirmed by the observations of localized plasticity patterns emergent at the linear stage of work hardening, which provide strong indications that the medium is separating into deforming and undeforming layers. The fruitful concept of self-organization has been proposed by [17], which states that a self-organizing system can attain spatial, temporal or functional inhomogeneity in the absence of any specific action from the outside. Note that the definition is used in a restricted, phenomenological, meaning; it implies no

concrete underlying mechanism responsible for the realization of self-organization process.

The concept of self-organization is frequently and successfully used for explanation of the formation of structure in active media studied in physics, chemistry, materials science or biology. The deforming medium can be similar to an active medium far from thermodynamic equilibrium in which the sources of energy are distributed over material volume.

We furnish strong evidence that the plastic flow also involves self-organization phenomena. According to [57], the generation of localized plastic flow autowaves causes a decrease in the entropy of the deforming system, which is the principal attribute of self-organization processes.

2.5 Autowave Equations

To offer adequate tools for describing autowave processes, a set of two equations has to be produced [45, 58] to describe the rate of change in the catalytic and damping factors. The justification of the choice of these two factors is far from to be trivial. By addressing plasticity, it is convenient to introduce plastic deformation, ε , and stress, σ , as catalytic and damping factors, respectively. Hence, equations for the rates $\dot{\varepsilon}$ and $\dot{\sigma}$ have to be derived on the base of general principles. The equation for $\dot{\varepsilon}$ is deduced from the condition of deformation flow continuity [43] as

$$\dot{\varepsilon} = \nabla \cdot (D_\varepsilon \nabla \varepsilon), \quad (6)$$

where the value D_ε is a transport coefficient and the term $D_\varepsilon \nabla \varepsilon$ is the deformation flow in the deformation gradient field $\nabla \varepsilon$; the coefficient D_ε depends on coordinates. By restricting our analysis to the case of uniaxial deformation along the axis x , we obtain

$$\partial \varepsilon / \partial t = \partial \varepsilon / \partial x \cdot \partial D_\varepsilon / \partial x + D_\varepsilon \partial^2 \varepsilon / \partial x^2 = f(\varepsilon) + D_\varepsilon \partial^2 \varepsilon / \partial x^2, \quad (7)$$

where $f(\varepsilon) = \partial \varepsilon / \partial x \cdot \partial D_\varepsilon / \partial x$ is a non-linear strain function.

The equation for $\dot{\sigma}$ can be derived from Euler's equation for hydrodynamic flow [59] as

$$\frac{\partial}{\partial t} \rho V_i = - \frac{\partial \Pi_{ik}}{\partial x_k}. \quad (8)$$

In the case of viscous medium, the momentum flux density tensor is given as $\Pi_{ik} = p \delta_{ik} + \rho V_i V_k - \sigma_{vis} = \sigma_{ik} - \rho V_i V_k$; the value δ_{ik} is unit tensor and V_i and V_k are flow rate components. The stress tensor $\sigma_{ik} = -p \delta_{ik} + \sigma_{vis}$ includes viscous stresses. By plastic deformation, $-p \delta_{ik} \equiv \sigma_{el}$; hence, $\sigma = \sigma_{el} + \sigma_{vis}$ or $\dot{\sigma} = \dot{\sigma}_{el} + \dot{\sigma}_{vis} = g(\sigma) + \dot{\sigma}_{vis}$.

The origination of viscous stresses, σ_{vis} , is due to plastic deformation inhomogeneity; the value σ_{vis} is related to variation in the rate of elastic waves propagating in the deforming medium, i.e. $\sigma_{vis} = \hat{\eta} \nabla V_t$. Here $\hat{\eta}$ is the dynamic viscosity of the medium and V_t is the propagation rate of transverse ultrasound waves. The equation $\sigma_{vis} = \hat{\eta} \nabla V_t$ can be written as $\partial \sigma_{vis} / \partial t = V_t \nabla \cdot (\hat{\eta} \nabla V_t) = \hat{\eta} V_t \partial^2 V_t / \partial x^2$. The value V_t depends on the acting stresses as $V_t = V_* + \varsigma \sigma$ [60]. Hence, equation for describing the rate of stress change may have the form analogous to that of Eq. (7), i.e. $\partial \sigma_{vis} / \partial t = \hat{\eta} V_t \partial^2 V_t / \partial x^2 = \hat{\eta} \varsigma V_t \partial^2 \sigma / \partial x^2$. Thus, we obtain

$$\dot{\sigma} = g(\sigma) + D_\sigma \partial^2 \sigma / \partial x^2, \quad (9)$$

where $D_\sigma = \hat{\eta} \varsigma V_t$ is the transport coefficient. It was shown in [60] that Eqs. (7) and (9) can be used to adequately describe plastic flow regimes.

2.6 On the Relation of Autowave Equations to Dislocation Theory

The problem of plasticity can be addressed in the frame of two different approaches, i.e. the autowave model proposed herein and dislocation theory. Of particular importance is the possible interrelation between the two approaches. Almost all the dislocation theories of plasticity are based on the Taylor-Orowan equation, which is used to describe the dislocation mechanism of plasticity [32], i.e.

$$\dot{\varepsilon} = b \rho_m V_{dist}, \quad (10)$$

where b is the Burgers vector and ρ_m is the density of dislocations moving at rate $V_{dist}(\sigma)$ under the action of applied stress. The first term in the right side of Eq. (7) is transformed by assuming that dislocation distribution is homogeneous and $d\varepsilon/dx \approx 1/s \cdot b/s \approx b\rho_m$. Here s is the distance between dislocations; the quantity b/s has the meaning of shear strain for dislocation path s and $s^{-2} \approx \rho_m$. For the case of plastic flow, it may be written $\eta_\varepsilon \approx L_{dist} \cdot V_{dist}$ (here $L_{dist} \approx \zeta x$ and $V_{dist} = const$ are, respectively, dislocation path and rate). Hence,

$$\dot{\varepsilon} = \zeta b \rho_m V_{dist} + D_\varepsilon \partial^2 \varepsilon / \partial x^2. \quad (11)$$

The right part of Eq. (11) accounts for two deformation flows, i.e. $\zeta b \rho_m V_{dist} \sim V_{dist}$ and $D_\varepsilon \partial^2 \varepsilon / \partial x^2$. The former flow is ‘hydrodynamic’ in character [59] and the latter is a ‘diffusion’ one.

Evidently, elimination of the second term will transform Eq. (11) into Eq. (10); hence, the Taylor-Orowan equation might be regarded as a special case of Eq. (11). Clearly, Eq. (10) finds limited use for plastic flow description, since it describes chaotic distributions of dislocations, which form no complex ensembles; hence,

Eq. (10) corresponds to work hardening due to long-range stress fields alone. Dislocation theory based on long-range stress fields might be called a linear one, while theory describing both flows from Eq. (11) might be regarded as an extended version of the dislocation theory.

A thorough analysis of Eq. (11) suggests that an appropriate dislocation model can be developed with the proviso that both terms in the right side of Eq. (11) are taken into account. In case the term $D_\epsilon \partial^2 \epsilon / \partial x^2$ is not eliminated from Eq. (11), it will initiate non-linear corrections of dislocation theory equations and thus expand significantly the area of application of dislocation theory. Such corrections might be of significance for crystals having high dislocation density.

Thus, the application of new experimental technique for plastic flow investigation enabled discovery of a new class of deformation phenomena—localized plastic deformation autowaves. These phenomena are addressed above, taking different but complementary approaches, i.e. autowave plasticity theory and dislocation theory. There is good reason to believe that compelling evidence has been provided for the existence of localized plasticity autowaves.

3 Elastic-Plastic Strain Invariant

The uniformity of localized plasticity phenomena observed for a wide range of materials suggests the existence of a general law for the localized plastic flow autowaves. This section focuses on the search for a quantitative relationship between the characteristics of elastic waves and autowaves.

3.1 Introduction of Elastic-Plastic Strain Invariant

We suggest a link between plastic flow macro-parameters and crystal lattice characteristics. For this purpose, two products are matched, i.e. λV_{aw} and χV_t , which characterize plastic flow and elastic deformation, respectively. The quantities χ and V_t are interplanar spacing of crystal lattice and transverse ultrasound wave velocity, respectively. Numerical analysis was performed using experimentally obtained values λ and V_{aw} as well as hand-book values χ and V_t . The data listed in Table 1 allow one to write the equality

$$\left\langle \frac{\lambda V_{aw}}{\chi V_t} \right\rangle = \hat{Z} \approx const. \quad (12)$$

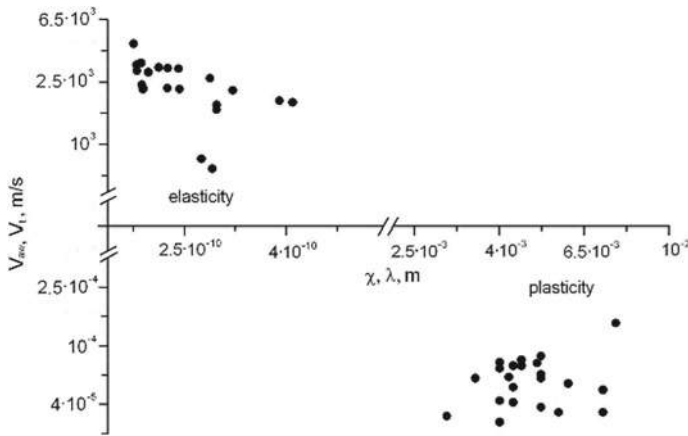
It can be seen from Fig. 4 that Eq. (12) holds true for all studied materials. The averaging of the value α was performed for seventeen materials to give $\langle \hat{Z} \rangle = 0.49 \pm 0.05 \approx 1/2 < 1$. This result constitutes both formal and physical proofs

Table 1 Data for verification of Eq. (12) for the elastic-plastic invariant of strain (data for λV_{aw} and χV_t are multiplied by $10^7 \text{ m}^2/\text{s}$)

	Metals											
	Cu	Zn	Al	Zr	Ti	V	Nb	γ -Fe	α -Fe	Ni	Co	Mo
λV_{aw}	3.6	3.7	7.9	3.7	2.5	2.8	1.8	2.55	2.2	2.1	3.0	1.2
χV_t	4.8	11.9	7.5	11.9	7.9	6.2	5.3	4.7	6.5	6.0	6.0	7.4
$\lambda V_{aw}/\chi V_t$	0.75	0.3	1.1	0.3	0.3	0.45	0.33	0.54	0.34	0.35	0.5	0.2

	Metals						
	Sn		Mg	Cd	In	Pb	Ta
$\lambda \cdot V_{aw}$	2.4		9.9	0.9	2.6	3.2	1.1
$\chi \cdot V_t$	5.3		15.8	3.5	2.2	2.0	4.7
$\lambda V_{aw}/\chi V_t$	0.65		0.63	0.2	1.2	1.6	0.2

	Alkali-halide crystals			Rocks	
	KCl	NaCl	LiF	Marble	Sandstone
$\lambda \cdot V_{aw}$	3.0	3.1	4.3	1.75	0.6
$\chi \cdot V_t$	7.0	7.5	8.8	3.7	1.5
$\lambda V_{aw}/\chi V_t$	0.43	0.4	0.5	0.5	0.4

**Fig. 4** Verification of invariant (12). The quantities χ , V_t (left) and λ , V_{aw} (right) are grouped in logarithmic coordinates in the neighborhood of average values

that the elastic and plastic processes involved in the deformation are closely related. Therefore, Eq. (12) has been labeled as *The elastic-plastic strain invariant*.

Apparently, $V_t \approx \chi \omega_D$ (here ω_D is the Debye frequency); hence, we write

$$\lambda V_{aw} \approx \hat{Z} \frac{V_t^2}{\omega_D} \approx \hat{Z} \frac{G}{\omega_D \rho} \approx \hat{Z} \frac{\partial^2 W / \partial v^2}{(\omega_D \chi) \rho} \approx \hat{Z} \frac{\partial^2 W / \partial v^2}{\xi_1}, \quad (13)$$

where $v \ll \chi$ is atomic displacement near interparticle potential minimum (W); the elastic modulus is expressed in terms of interparticle potential as $G = \chi^{-1} \partial^2 W / \partial v^2$ [61] and the value $\xi_1 = (\omega_D \chi) \rho = V_t \rho$ is specific acoustic resistance of the medium, which is shown to be related to crystal lattice perturbation due to dislocation motion [28]. The interparticle potential from Eq. (13) is

$$W(v) \approx \frac{1}{2} \cdot (\partial^2 W / \partial v^2) v^2 + \frac{1}{6} (\partial^3 W / \partial v^3) v^3 = \frac{1}{2} p v^2 - \frac{1}{3} q v^3, \quad (14)$$

where p is the coefficient of quasi-elastic coupling and q is anharmonicity coefficient. With the proviso that $\frac{1}{2} p \cdot v^2 \gg |-\frac{1}{3} q v^3|$, Eq. (13) assumes the form

$$\lambda V_{aw} \approx p / \xi_1 \approx p / V_t \rho, \quad (15)$$

where λV_{aw} can be taken as a criterion of plasticity [29].

3.2 Generalization of Elastic-Plastic Strain Invariant

The criterion λV_{aw} from Eq. (15) also holds good for deformation initiated by chaotically distributed dislocations. Let mobile dislocation density be ρ_m ; then the average distance between dislocations, which is equal to the dislocation path, is given as $\langle s \rangle = \rho_m^{-1/2}$. According to [32], $\sigma \approx (Gb/2\pi) \rho_m^{1/2}$; hence, we can write $\rho_m^{-1/2} = \langle s \rangle = Gb/2\pi \sigma \sim \sigma^{-1}$. The rate of quasi-viscous motion of dislocations is $V_{disl} = (b/B) \cdot \sigma$. Here B is the coefficient of dislocation drag by the phonon and electron gases [62]. Hence,

$$l \cdot V_{disl} \approx const = \frac{Gb^2}{2\pi \cdot B}. \quad (16)$$

Using the values $G \approx 40$ GPa and $B \approx 10^{-4}$ Pa s, which are conventionally employed for dislocation motion descriptions, we obtain $l V_{disl} \approx 10^{-6}$ m²/s. The latter value is close to the calculated value of the product $\hat{Z} \chi V_t$ obtained for studied materials (see Table 1).

The above suggests that we have established a reliable quantitative criterion for analyzing the interaction between the elastic deformation, which occurs on the micro-scale level, and the macro-scale plastic deformation. This criterion, in its universal form, applies to autowaves in question as well as to elastic and plastic deformation via dislocation motion. Therefore, this criterion is considered as a more general form of the elastic-plastic strain invariant:

$$\lambda V_{aw} = l V_{disl} = \hat{Z} \chi V_t \approx 10^{-6} \text{ m}^2/\text{s}. \quad (17)$$

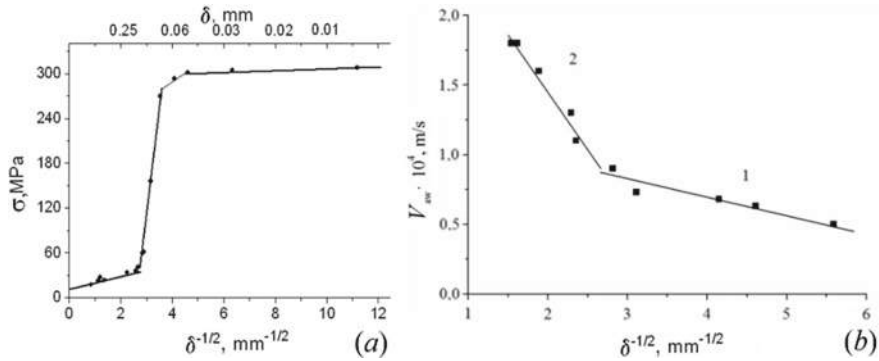


Fig. 5 Characteristics of autowaves obtained for polycrystalline Al: **a** strength limit as a function of grain size; **b** autowave rate as a function of grain size (sections 1 and 2 correspond, respectively, to the ranges $0.005 \leq \delta \leq 0.15$ mm and $0.15 \leq \delta \leq 5$ mm)

To provide a framework for validating the proposed strain invariant, a special-purpose series of experiments were carried on for polycrystalline aluminum samples having grain sizes in the range $5 \times 10^{-3} \leq \delta \leq 5$ mm. We obtained grain size dependencies of strength limit, $\sigma_B(\delta)$. The Hall-Petch relation $\sigma_B(\delta) = \sigma_0 + k_B \delta^{-1/2}$ [63] has been plotted; it is illustrated in Fig. 5a. It can be seen that for $\delta_0 \approx 0.1 \dots 0.2$ mm, a jump-wise variation occurs in the value $\sigma_B(\delta)$. The dependence $V_{aw}(\delta)$ demonstrated in Fig. 5b has a similar form. The value V_{aw} would also vary over the entire range of grain sizes; however, the surprising thing is that the ratio $\lambda V_{aw} / \chi V_t \approx 1/2$ would remain constant for both $\delta > \delta_0$ and $\delta < \delta_0$.

In view of the above, invariant (12) shows promise for gaining an insight into the nature of localized plastic flow. All the basic regularities of localized plastic flow autowaves can be deduced from Eq. (12). Therefore, the invariant is expected to play an important role in the development of new notions of plasticity.

3.3 On the Strain Invariant and Autowave Equations

Now certain theoretical considerations concerning the origin of invariant (12) will be explored. Particular emphasis is placed upon the fact that the quantities λV_{aw} and χV_t from Eq. (12) have evidently the dimension $L^2 T^{-1}$. Note that in the right sides of Eqs. (7) and (9) the coefficients D_ε and D_σ for the terms containing second-order derivatives $\partial^2 / \partial x^2$ have the same dimension.

In view of the above, we assume that

$$\lambda V_{aw} \equiv D_\varepsilon \quad (18)$$

and

$$\chi V_t \equiv D_\sigma. \quad (19)$$

The above identification is valid, considering the dimensions of the quantities λV_{aw} and χV_t ; moreover, Eq. (9) contains expression for the coefficient D_σ in which velocity V_t appears. According to Eq. (12), $\widehat{Z} < 1$, i.e. $D_\sigma > D_\varepsilon$. Hence, the condition required for the autowave generation is satisfied [45].

Given their dimensions, these quantities may be either diffusion coefficients or kinematic viscosities of a media. In what follows, the reasoning behind the latter suggestion is formulated. Using the dimensionality analysis, we write $\lambda \approx \sqrt{2\eta \cdot t}$; here the value $\eta \approx 10^{-6} \text{ m}^2/\text{s}$ and the value $t \approx T_{aw} \approx 10^2 \text{ s}$. Apparently, $\lambda \approx 10^{-2} \text{ m}$, which is identical to the autowave length.

3.4 Some Consequences of the Strain Invariant

Implicitly, it is generally assumed that the total deformation is a sum of elastic and plastic strains, i.e. $\varepsilon_{tot} = \varepsilon_e + \varepsilon_p$. By virtue of $\varepsilon_e \ll \varepsilon_p$ and $\varepsilon_{tot} \approx \varepsilon_p$, the contribution of elastic strain is frequently neglected altogether. However, invariant (12) implies that the quantities ε_e and ε_p are closely related; hence, these quantities should be taken properly into account by addressing plastic flow localization. A model for simulation of plastic deformation should be based on the fundamental assumption that plastic form changing involves both elastic and plastic deformation mechanisms that are interdependent in principle.

To provide arguments in favor of the strain invariant, its consequences have been analyzed. It is found that the regularities of plastic flow localization, which are given by Eqs. (3) through (5), can be derived from Eq. (12) as well. Consider the respective procedures step by step.

It follows from Eq. (3) that the propagation rate of localized plasticity autowave is inversely proportional to the work hardening coefficient. To prove it, we differentiate Eq. (12) with respect to deformation ε

$$\lambda \frac{dV_{aw}}{d\varepsilon} + V_{aw} \frac{d\lambda}{d\varepsilon} = \widehat{Z} \chi \frac{dV_t}{d\varepsilon} + \widehat{Z} V_t \frac{d\chi}{d\varepsilon}. \quad (20)$$

Hence,

$$V_{aw} = \left(\frac{d\lambda}{d\varepsilon} \right)^{-1} \left(\widehat{Z} \chi \frac{dV_t}{d\varepsilon} + \widehat{Z} V_t \frac{d\chi}{d\varepsilon} - \lambda \frac{dV_{aw}}{d\varepsilon} \right). \quad (21)$$

Since the interplanar spacing of crystal is independent of plastic deformation, in Eq. (21) $\widehat{Z} V_t \frac{d\chi}{d\varepsilon} \approx 0$. Hence,

$$V_{aw} = \widehat{Z} \chi \frac{dV_t}{d\lambda} - \lambda \frac{dV_{aw}}{d\lambda}. \quad (22)$$

Further we shall refer to [32] who reasoned that work hardening coefficient can be expressed as a ratio of two parameters of the deforming medium structure, e.g. $\chi \ll \lambda$, i.e. $\theta \approx \chi/\lambda$ and $dV_{aw}/d\lambda < 0$. Thus, rearrangement of Eq. (22) yields the same result as the experiment does, i.e.

$$V_{aw} = \widehat{Z}\chi \frac{dV_t}{d\lambda} - \chi \frac{dV_{aw}}{d\lambda} \frac{\lambda}{\chi} \approx V_0 + \frac{\Xi}{\theta}. \quad (23)$$

Equation (4) also follows from Eq. (12), which can be rewritten as

$$V_{aw} = \frac{\Theta}{\lambda} = \frac{\Theta}{2\pi}k, \quad (24)$$

where $\Theta = \chi V_t/2$. If $V_{aw} = d\omega/dk$, then $d\omega = (\Theta/2\pi) \cdot k \cdot dk$. Integration of the latter equality is performed:

$$\int_{\omega_0}^{\omega} d\omega = \frac{\Theta}{2\pi} \int_0^{k-k_0} k dk \quad (25)$$

to yield dispersion law of quadratic form

$$\omega = \omega_0 + \frac{\Theta}{4\pi}(k - k_0)^2, \quad (26)$$

which is equivalent to Eq. (4) with the proviso that $\alpha = \Theta/4\pi$.

The grain size dependence of autowave length, $\lambda(\delta)$, given by Eq. (5) also follows from invariant (12). Indeed, Eq. (12) can be rewritten as

$$\lambda = \widehat{Z}\chi \frac{V_t}{V_{aw}}. \quad (27)$$

By virtue of the fact that the quantities V_t and V_{aw} depend on grain size, δ , [54, 64], differentiation of Eq. (12) is performed with respect to δ as

$$\frac{d\lambda}{d\delta} = \widehat{Z}\chi \frac{d}{d\delta} \left(\frac{V_t}{V_{aw}} \right) = \widehat{Z}\chi \left(\frac{V_{aw}dV_t/d\delta - V_t dV_{aw}/d\delta}{V_{aw}^2} \right). \quad (28)$$

With the proviso that $V_{aw} = \alpha\chi V_t/\lambda$, Eq. (28) can be rewritten as

$$d\lambda = \widehat{Z}\chi \left(\frac{dV_t}{d\delta V_{aw}} - V_t \frac{dV_{aw}}{d\delta V_{aw}^2} \right) d\delta = \left(\frac{dV_t}{d\delta V_t} \lambda - \frac{1}{\alpha\chi V_t} \frac{dV_{aw}}{d\delta} \lambda^2 \right) d\delta, \quad (29)$$

or

$$d\lambda = (a_1\lambda - a_2\lambda^2)d\delta. \quad (30)$$

A solution of Eq. (30) yields Eq. (5). Taking into account Eq. (29), the coefficients of Eq. (5) take on the meaning: $a_1 = \frac{dV_t}{V_t d\delta} = \frac{d \ln V_t}{d\delta}$ and $a_2 = \frac{2dV_{aw}}{\chi V_t d\delta}$.

Thus, Eqs. (3)–(5) follow from elastic-plastic invariant (12) and depend on the lattice properties of the deforming medium. It can thus be concluded that the elastic and plastic processes occurring in the deforming solid are closely related.

Plastic flow occurring in a medium is described by Eq. (7), which can be derived from invariant (12) as well. To do this, Eq. (12) is rewritten as

$$\frac{\lambda}{\chi} = \hat{Z} \frac{V_t}{V_{aw}}, \quad (31)$$

where the term $\lambda/\chi \equiv \varepsilon$ is assumed to be deformation. By applying the differential operator $\partial/\partial t = D_\varepsilon \partial^2/\partial x^2$ to the right and left sides of Eq. (31), we obtain

$$\frac{\partial \varepsilon}{\partial t} = \hat{Z} D_\varepsilon \frac{\partial^2}{\partial x^2} (V_t/V_{aw}). \quad (32)$$

Differentiation of Eq. (32) yields

$$\frac{\partial \varepsilon}{\partial t} = \hat{Z} D_\varepsilon \left(-V_t \frac{\partial^2 V_{aw}^{-1}}{\partial x^2} + V_{aw}^{-1} \frac{\partial^2 V_t}{\partial x^2} \right) \quad (33)$$

According to [56], the ultrasound rate V_t varies in an intricate fashion in the deforming medium, while the autowave propagates at a constant rate at the linear stage of work hardening. Thus, reduction of Eq. (33) yields

$$\frac{\partial \varepsilon}{\partial t} = -\hat{Z} D_\varepsilon V_t \frac{\partial^2 V_{aw}^{-1}}{\partial x^2} + D_\varepsilon \frac{\partial^2 \varepsilon}{\partial x^2} = f(\varepsilon) + D_\varepsilon \frac{\partial^2 \varepsilon}{\partial x^2}, \quad (34)$$

which is equivalent to Eq. (7).

Thus, the main features of localized plastic flow are described by Eqs. (7) through (9), which are derived from invariant (12) obtained on the base of experimental evidence. Equation (12) states that the characteristics of localized plastic flow are determined by the lattice characteristics of the medium. This thesis in the form of logical implication would assert that the processes involved in elastic and plastic deformation in solids are closely related. It must be admitted that among the factors governing the processes of elastic deformation are not only the crystal lattice properties, but also its interparticle potential, whereas the processes of plastic deformation are governed by the behavior of lattice defects alone.

4 The Model of Localized Plastic Flow

By addressing the localization behavior of plastic deformation in solids, the acoustic characteristics manifested by the deforming medium also call for further investigation. Up to now, the acoustic characteristics of the deforming medium were addressed in terms of energy dissipation, in particular, in internal friction studies and related problems. In what follows, this subject is discussed at greater length.

4.1 *Plastic and Acoustic Characteristics of the Deforming Medium*

We are dealing here with the rate of transverse elastic waves, V_t , which appears in Eq. (12) for strain invariant and in the expression of acoustic resistance of the medium, which enters into Eq. (13). It is also shown above that the coefficient Ξ from Eq. (3), which describes the autowave propagation rate, depends on the rate of transverse elastic waves [60].

One must take into account another characteristic of the deforming medium, i.e. phonon gas viscosity [65] which appears in Eq. (16). At first glance, this quantity seems to be out of place in the analysis of slow processes of localized plasticity autowave propagation. This value is generally determined for a sample under impact loading by addressing high rate motion of dislocations. Nonetheless, phonon gas viscosity made its appearance in our discussion from the following considerations. The development of plastic deformation is due to dislocation motion. Dislocations move over the local obstacles [32], the rate of dislocation motion is given as $V_{disl} \approx (b/B)\sigma$ and is controlled by the phonon gas (see above). The appearance of the quantity B in the latter relation is accounted for by the occurrence of moving dislocations within the localized plasticity nuclei.

The mechanical and acoustic characteristics of the deforming medium are found to be closely related [28]. This finding is supported by the experimental evidence, which strongly suggests that acoustic processes play an important role in the development of localized plastic flow. Available acoustic emission data suggests that structural inhomogeneities would emerge in the deforming medium due to a traveling deformation front. Thus, the acoustic emission sources occurring in material bulk have to be linked to the localized plastic flow nuclei emergent in the deforming solid [56]. To address the nature of localized plasticity, a two-component model has been formulated, in which a key role is assigned to the acoustic properties of the deforming solid. Acoustic emission pulses play the role of information system and control the dynamics of form changing.

4.2 Two-Component Model of Localized Plasticity

In the conceptual framework used to address the basic problem of autowave formation is the nature of self-organization, which manifests itself in the deforming medium as a spontaneous emergence of autowave structure. Physical interpretation of Eqs. (7) and (9) might prove productive for elucidation of the problem. Kadomtsev [49] advanced the idea that a self-organizing system will separate spontaneously into dynamic and information subsystems, interacting with one another.

The idea of the proposed model is as follows. In the course of plastic deformation local stress concentrators would form and disintegrate; these are considered as slowed-down shears. Elementary stress relaxation act is due to breaking from a local obstacle, which involves acoustic emission [66]. These acoustic signals will activate other stress concentrators, to so that the same process is repeated over and over again. Thus acoustic emission signals propagating in the deforming medium play the role of information subsystem; dislocation shears are involved in the plastic deformation proper and operate as a dynamic subsystem. The model developed is made up of two components: acoustic emission and dislocation mechanisms of plasticity, which have been studied sufficiently, although in different contexts. The generation of acoustic signals was considered in connection with the initiation of dislocation shears, while the reverse process, i.e. initiation of shears due to acoustic pulses, has not been touched on thus far.

In what follows, the performance of the proposed model is assessed. Acoustic signal can propagate in non-uniform dislocation substructure, which forms by deformation and is observable by transmission electron microscopy, e.g. dislocation cell having size $R \approx 0.01$ mm. It is proposed by [56] that such cell be regarded as acoustic lens, which has focal length, f_l , given as

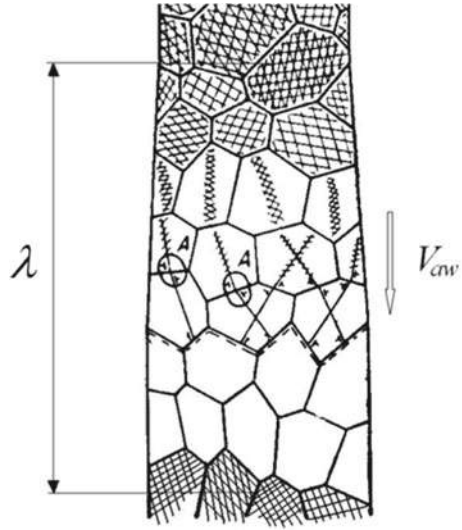
$$f_l \approx \frac{R}{V_t^{(def)} / V_t - 1}, \quad (35)$$

where the ratio of ultrasound rates, $V_t^{(def)} / V_t$, observed for non-deformed and deformed volumes plays the role of acoustic refractive index. The initiation of plastic deformation is due to the ultrasound waves focusing at distance $\lambda \approx f_l \approx 10^{-2}$ m from the active localized plasticity nucleus.

Now let's estimate acoustic emission pulses in terms of energy expenditure required for activating dislocation shears. According to [67], the time needed for dislocations to move over barriers during thermally activated motion, is $\tau \approx \exp\left(\frac{U_0 - \gamma\sigma}{k_B T}\right)$. Here $U_0 - \gamma\sigma = H$ is the process enthalpy and k_B is the Boltzmann constant. Generally, $H \approx 1$ eV and $\tau \approx 10^{-6}$ s. For the case of $H = U_0 - \gamma\sigma - \varepsilon_{ph}$, where the phonon energy is given as $\varepsilon_{ph} = \hbar\omega_D \approx 0.3$ eV, $\tau \approx 5 \times 10^{-7}$ s. The generation of a new front is presented schematically in Fig. 6.

It is an established fact that the perfect crystal lattice is a source of crystal defects responsible for plastic form changing; therefore, its properties must be taken into account by addressing self-organization processes as well. Hence, the basic premise

Fig. 6 Scheme of autowave formation



of the given paper is that the regular features of plastic flow macrolocalization are directly related to the lattice characteristics.

5 Plastic Flow Viewed as a Macroscopic Quantum Phenomenon

An innovative approach to the plasticity problem can be developed using elastic-plastic strain invariant (12), which has a deep physical meaning.

5.1 Localized Plastic Flow Autowaves and the Planck Constant

In the course of plastic flow the autowave processes are generated in the deforming medium. This mechanism has been established experimentally for all the plastic flow stages. The findings are giving us a clue to the most distinctive features of the plastic flow and thus provide additional insights into basic plasticity problems. Clearly, the next step in the development of this model is a quantum representation of the plastic flow. This necessitates introduction of a quasi-particle corresponding to the localized plastic flow autowave. The strong evidence that lends support to this idea is considered below.

Numerical analysis was performed for the experimental data on λ and V_{aw} , which produced an unexpected result. Thus it is found that the product $\lambda V_{aw} \rho r_{ion}^3$ (here ρ

Table 2 Estimation of the Planck constant with the help of Eq. (36)

	Metals										
	Cu	Zn	Al	Zr	Ti	V	Nb	γ-Fe	α-Fe	Ni	Co
$h \cdot 10^{34}$	11.9	9.3	2.8	6.1	4.9	3.5	4.9	4.6	4.6	6.1	7.1

	Metals							
	Sn	Mg	Cd	In	Pb	Ta	Mo	Hf
$h \cdot 10^{34}$	8.9	4.9	7.4	9.9	18.4	5.5	3.0	7.3

is metal density and $r_{ion} \approx \chi$ is ion radius of metal) is close to the Planck constant $h = 6.63 \times 10^{-34}$ J s [68]. Hence, we write

$$\lambda V_{aw} \rho r_{ion}^3 \approx h.$$

(36)

The validity of Eq. (36) is justified by the data listed in Table 2 (note: handbook values of ion radii are used herein). On the base of these data the average value of the Planck constant was calculated for thirteen metals; the resultant value $\langle h \rangle = (6.9 \pm 0.45) \times 10^{-34}$ J s, with the ratio $\langle h \rangle / h = 1.04 \pm 0.06 \approx 1$, i.e. $\langle h \rangle = h$.

Using a standard statistical procedure [69], the quantities $\langle h \rangle$ and h were matched. Let the value $\langle h \rangle$ be defined as the average of thirteen measurements ($n_1 = 19$). On the other hand, we operated on the premise that the value h was determined in a single measurement ($n_2 = 1$) in the absence of dispersion. The statistical significance of coincidence of the quantities $\langle h \rangle$ and h was determined with 95% confidence level by Student’s t -test as

$$t = \frac{\langle h \rangle - h}{\hat{\sigma}} \cdot \sqrt{\frac{n_1 n_2}{n_1 + n_2}},$$

(37)

where the value $\hat{\sigma}$ is the square root of the overall estimate of dispersion. This procedure shows that the values $\langle h \rangle$ and h are statistically identical, i.e. $\langle h \rangle = h$. It is pertinent to note that h is the fundamental constant; hence, its appearance in Eq. (36) is in no way accidental—this suggests that plastic deformation physics is related to quantum mechanics.

5.2 Introduction of a New Quasi-particle and Its Applications

Further on, the plasticity problem is approached using quantum ideas. By further elaborating the autowave concept of localized plasticity, we were guided by the fundamentals of modern condensed-state physics where quasi-particle concept is generally introduced and freely employed for simplifying description of solids [70]. Our consideration of localized plastic flow autowaves is based on the concept of wave-particle dualism.

In the first place, the mass of the hypothetical quasi-particle is of principal importance. Clearly, the characteristics of the quasi-particle have to be related to those of the autowave. Thus the quasi-particle mass is defined as follows. On the base of data obtained for dispersion autowaves in Al and γ -Fe, we write

$$m_{ef} = \left(\frac{d^2 U}{dp^2} \right)^{-1} = \hbar \frac{d^2}{dk^2} [\omega(k)], \quad (38)$$

where the values U and p denote, respectively, the energy and quasi-momentum of the quasi-particle. Another way of looking at it is proposed by [71, 72]. Thus the de Broglie formula can be employed to address localized plastic flow autowaves as

$$m_{ef} = \frac{h}{\lambda V_{aw}}. \quad (39)$$

An alternative method is the use of the term from Eq. (36)

$$\rho r_{ion}^3 = m_{ef}. \quad (40)$$

Apparently, the effective mass of a quasi-particle having size $\sim r_{ion}$ can also be defined from Eqs. (39) and (40).

Using Eqs. (38) through (40), the mass of the quasi-particle was found; the values obtained are in the range $0.5 \leq m_{ef} \leq 1.5$ a.m.u. (atomic mass unit). The averaged value $\langle m_{ef} \rangle \approx 1$ a.m.u. is a rough estimate of the quantity m_{ef} .

The hypothetic quasi-particle is named *auto-localizon*. The next task is equating the propagation rates of the autolocalizon and the autowave. The mobility of autolocalizon is affected by the phonon and electron gases in solids. Thus the effective mass of the auto-localizon is regarded as its virtual mass, which is defined by the resistance of both gases to the motion of auto-localizon. Strong evidence was recently obtained in support of this conjecture. The effective mass was calculated from Eq. (38) for a range of metals.

We will now look at some possibilities offered by this approach. Using Eq. (38), the formula for elastic-plastic strain invariant (12) can be rewritten as

$$\frac{h}{\lambda V_{aw}} \approx \widehat{Z}^{-1} \frac{h}{\chi V_t}, \quad (41)$$

where $h/\chi V_t = m_{ph}$ and $h/\lambda V_{aw} = m_{a-l}$ are, respectively, the phonon and the auto-localizon masses. Evidently, Eq. (41) is equipotent to the equality $m_{a-l} \approx \widehat{Z}^{-1} m_{ph}$. Hence, Eq. (41) accounts for the mechanism, which is responsible for the generation of dislocations due to phonon condensation [73].

This concept is elaborated in the frame of conventional approach adopted in the solids physics, which involves introduction of a quasi-particle for description of wave processes. By way of an example, it well suffices to mention elementary

excitations in media [70]. One of the first attempts of this kind was made by [74] who introduced a quasi-particle named *crackon* in order to address mechanisms of brittle crack propagation.

It should be reminded that such ideas have long been in the air; therefore, attempts at introducing quantum concepts into the physics of plasticity are by no means scarce. Thus the quantum tunneling effect was used by [2, 75, 76] in descriptions of low-temperature processes involving dislocations breakaway from pinning points. Later on, [77] supplied a detailed explanation of this phenomenon for the case of dislocation motion in the Peierls-Nabarro potential relief. Steverding [78] made use of quantization of elastic waves propagating by material fracture. Zhurkov [79] introduced the notion of elemental excitation in crystals, which was termed as *dilaton*. Later on, the possible existence of a specific precursor of deformation or fracture was hypothesized by [80] who coined the name *frustron* for this phenomenon. It has been shown that an elementary act of interatomic bond rupture has activation volume close to that of an atom. Evidently, ideas of this kind are transparent enough; the underlying theoretical premises are based on the discreteness of crystal lattice in which generation and evolution of elementary acts of plasticity takes place. The autowave and quasi-particle concepts are distinct, though complementary and interrelated approaches.

In the frame of quasi-particle concept, the length of localized plasticity autowave can be estimated. With this aim in view, the motion of autolocalizon in the phonon gas is considered. In view of the fluctuations of phonon gas density, it is proposed that the autolocalizon be involved in the Brownian motion. In accordance with Einstein's theory, the free path of the Brownian particle is

$$s \approx \sqrt{\frac{k_B T}{\pi \hat{\eta} r_{a-l}}} t, \quad (42)$$

where $\hat{\eta}$ is the dynamic viscosity of the phonon gas and t is the time given as $t \approx 2\pi/\omega$ (here ω is the frequency of localized plastic flow autowave).

Hence, the free path of the quasi-particle is presented as autowave length λ and is given as follows. Assume that $T = 300$ K; the autolocalizon has size $r_{a-l} \approx 10^{-10}$ m; the autowave period is $t \approx 10^3$ s and $\hat{\eta} \approx 10^{-4}$ Pa s (the latter value was obtained by [62] in high-velocity dislocation motion tests). The resultant value $s \approx 10^{-2}$ m, which is evidently close to the autowave length, λ . Thus the application of Eqs. (35) and (42) yields equivalent numerical estimates.

5.3 Plasticity Viewed as a Macro-scale Quantum Phenomenon

A close relation has been established between the deformation and acoustic characteristics of the deforming medium, which suggests that the deformation processes can be described by a hybridized excitation spectrum (Fig. 7). Such a spectrum is

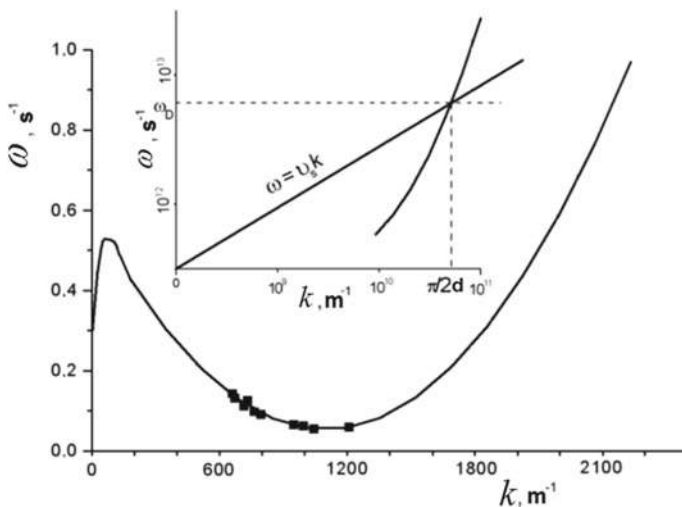


Fig. 7 Generalized dispersion curve obtained for elastically and plastically deforming solids (insert similar dependencies obtained for high-frequency oscillation spectrum)

obtained by the imposition of the linear dispersion relation $\omega \approx V_l k$ for elastic waves and quadratic the same $\omega = \omega_0 + \alpha(k - k_0)^2$ for localized plasticity autowaves.

For validation the coordinates $\hat{\omega}$ and \hat{k} were estimated for the point of intersection of the plots in a high-frequency spectral region. The frequency $\hat{\omega} \approx \omega_D$ and the wave number \hat{k} correspond to the minimal length of elastic wave, which is of the order of distance between close packed planes, i.e. $\hat{k} \approx 2\pi/\chi$. The above evidence indicates that the generalized dispersion relation holds good for both the phonons and the auto-localizations.

The dispersion curve illustrated in Fig. 7 is suggestive of a remarkable analogy with the dispersion relation obtained for superfluid ^4He [70, 81]. The latter dispersion relation shows a minimum corresponding to the origination of ‘*rotons*’, i.e. quasi-particles having effective mass $m_{rot} \approx 0.64$ a.m.u.; moreover, the quadratic dispersion law obtained for rotons is similar to Eq. (4). The above analogy is indicative of a similarity of localized plasticity and superfluidity. Whether this is a formal similarity or whether it has a physical meaning, remains to be seen.

Additional argument in favor of this attractive conjecture is as follows. The superfluidity of ^4He is attributed to the occurrence of normal and superfluid components in liquid ^4He at $T \leq 2.17$ K; the respective dynamic viscosities obtained for these components are $\hat{\eta}_n$ and $\hat{\eta}_{sf} \ll \hat{\eta}_n$. The plastic flow occurring in the deforming medium would involve both slow motion of individual material volumes, which undergo form changing, and fast dislocation motion. Slow process corresponds to high viscosity of material, $\hat{\eta}_{mat} = G\tau \approx 10^{10}$ Pa s (here $\tau \approx 10^3$ s is a characteristic time). The velocity of dislocation motion, $V_{disl} \approx (b/B) \cdot \sigma$, is controlled by the phonon gas viscosity, $B \approx 10^{-4}$ Pa s [62], i.e. $\hat{\eta}_{mat}/B \approx 10^{14}$.

Table 3 The macroscopic quantum phenomena

Phenomenon	Observable quantum macrocharacteristic	
Superconductivity	Magnetic flux	$\Phi = \frac{\pi \hbar c}{e} \cdot n$
Superfluidity	Rotation rate of ^4He	$v_s = \frac{\hbar}{A} \cdot \frac{1}{r} \cdot n$
The Hall quantum effect	Hall's resistance	$R_H = \frac{h}{e^2} \cdot \frac{1}{n}$
Localized plastic flow	Magnitude of deformation jump	$\delta L \approx \frac{h}{\rho \chi^3 V_{aw}} \cdot m \approx \frac{h}{m_{a-l} V_{aw}} \cdot n$

$\hbar = h/2\pi$ is the Planck constant; c —velocity of light; e —electron charge; A —atomic mass of ^4He ; r —radius; $m = 1, 2, 3 \dots$

Three macroscopic quantum effects are well-known in physics, i.e. superconductivity, superfluidity and the quantum Hall effect [82]. The characteristics of these effects are listed in Table 3. On the base of data obtained in this study, the localized plasticity phenomenon might be included in the same ‘short list’. In what follows, we shall attempt to apply a quasi-particle approach to the analysis of serrated plastic deformation similar to the Portevin-Le Chatelier effect [83–85]. Assume that autowaves having length λ are arranged along sample length, L . Then the number of autowaves is given as $\lambda = L/n$ where $n = 1, 2, 3 \dots$. The deformed sample has length $L \approx L_0 + \delta L$ (here L_0 is initial length); hence, $\delta L \approx \lambda$. Thus from Eq. (35) follows

$$\delta L \approx \frac{h}{\rho \chi^3 V_{aw}} n \approx \frac{h}{m_{a-l} V_{aw}} n. \quad (43)$$

The autolocalizon mass $m_{a-l} \approx \rho \chi^3 \approx \rho r_{ion}^3$ appears in Eq. (43) which states that a jump-wise elongation of the tensile sample is necessary, i.e. $\delta L \sim n$. For the linear work hardening stage, $V_{aw} = \text{const}$; hence, $\frac{h}{\rho \chi^3 V_{aw}} = \text{const}$. Given sufficient instrumentation sensitivity, the recorded curves $\sigma(\varepsilon)$ will invariably exhibit a serrated behavior; moreover, accommodation of the sample length will occur to fit the general autowave pattern. Numerical estimates were made which suggest that for $n = 1$, $\rho \approx 5 \times 10^3 \text{ kg/m}^3$ and $\chi \approx 3 \times 10^{-10} \text{ m}$ hence, the elongation jump $\delta L_{m=1} \approx 10^{-4} \text{ m}$. For the sample length $L \approx 10^{-1} \text{ m}$ the elongation jump corresponds to the deformation jump $\delta \varepsilon_{m=1} \approx 10^{-3}$, which is a close match of the experimentally obtained value.

It also follows from Eq. (43) that an increase in the loading rate would cause a decrease in the deformation jump value, i.e. $\delta L \sim V_{aw}^{-1}$. This inference is supported by the experimental results obtained for Al samples tested at 1.4 K at different loading rates [86]. Thus, the autowave rate was found to be proportional to the motion velocity of the testing machine crossheads, i.e. $V_{aw} \sim V_{mach}$. According to Eq. (43), the velocity V_{aw} will increase with rate V_{mach} , while the deformation jumps will grow smaller.

6 Conclusions

1. The discussion of factual evidence cited herein enables formulation of a new idea of the nature of plasticity. In the frame of proposed concept, the plastic flow localization is due not only to the formation and to redistribution of defects (dislocations) in the deforming medium, but also to the lattice and material characteristics related to quantum mechanics. It is found that the parameters of plastic flow localization are related to the quantities h , χ and V_I . This relation is a qualitative one, while material structure plays a subordinate, quantitative role.
2. An analysis of the plastic flow suggests that regular features, which are manifested in all deforming solids, distinguish the deformation process. The kinetics of plastic flow is determined by a regular changeover in the localization patterns (autowave modes).
3. Elastic-plastic strain invariant is introduced to relate the processes involved in plastic and elastic deformation. It is shown that the main laws of autowave plastic deformation are corollaries of this invariant.
4. A well-founded conjecture is proposed that the localized plasticity phenomenon belongs to the category of quantum effects manifested on the macro-scale level. Validation of this hypothesis is also provided. To advance this idea, a quasi-particle of localized plastic deformation (auto-localizon) is introduced.

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Three-Component Wear-Resistant PEEK-Based Composites Filled with PTFE and MoS₂: Composition Optimization, Structure Homogenization, and Self-lubricating Effect



Sergey V. Panin, Lyudmila A. Kornienko, Nguyen Duc Anh,
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Abstract The aim of this work was to design and optimize compositions of three-component composites based on polyetheretherketone (PEEK) with enhanced tribological and mechanical properties. Initially, two-component PEEK-based composites loaded with molybdenum disulfide (MoS₂) and polytetrafluoroethylene (PTFE) were investigated. It was shown that an increase in dry friction mode tribological characteristics in metal-polymer and ceramic-polymer tribological contacts was attained by loading with lubricant fluoroplastic particles. In addition, molybdenum disulfide homogenized permolecular structure and improved matrix strength properties. After that, a methodology for identifying composition of multicomponent PEEK-based composites having prescribed properties which based on a limited amount of experimental data was proposed and implemented. It was shown that wear rate of the “PEEK + 10% PTFE + 0.5% MoS₂” composite decreased by 39 times when tested on the metal counterpart, and 15 times on the ceramic one compared with neat PEEK. However, in absolute terms, wear rate of the three-component composite on the metal counterpart was 1.5 times higher than on the ceramic one. A three-fold increase in wear resistance during friction on both the metal and ceramic counterparts was achieved for the “PEEK + 10% PTFE + 0.5% MoS₂” three-component composite compared with the “PEEK + 10% PTFE”. Simultaneous loading with two types of fillers slightly deteriorated the polymer composite structure compared with neat PEEK. However, wear rate was many times reduced due to facilitation of transfer film formation. For this reason, there was no microabrasive wear on both metal and ceramic counterpart surfaces.

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1 Introduction

Polyetheretherketone (PEEK) is one of the prospective structural polymeric materials due to the unique combination of operational characteristics: high strength and toughness, thermal and chemical resistance, as well as biocompatibility. PEEK is also stable during long-term operation at low and elevated temperatures (from -40 to 260 °C) while maintaining high mechanical properties. In addition, PEEK has enough melt flow rate, which facilitates its processing and application, including in additive manufacturing of complex-shaped parts [1, 2].

Varying composition of fillers changes PEEK characteristics and expands application areas. In particular, reinforcing fibers (carbon, glass, aramid, etc.) are loaded to increase its mechanical properties [3–5]. PEEK-based composites containing about 30 wt% carbon or glass fibers are most widely used as a polymer structural material [6, 7]. However, as has been shown in [8], metal counterparts wear out rapidly even during friction on neat PEEK. If PEEK has been loaded with reinforcing fibers, wear rate increases many times [9].

Traditionally, the problem of low PEEK antifriction properties has been solved by loading with solid lubricant fillers. One of the most common among them is polytetrafluoroethylene (PTFE), which in some cases reduce PEEK wear rate by several orders of magnitude [10–14]. Recently, PEEK-based nanocomposites have been actively designed as well [15, 16]. Meanwhile, some published data on effect of the fillers on PEEK-based composite wear resistance during dry sliding friction have been controversial [17, 18]. Nevertheless, loading with (nano)particles of various compositions as solid lubricant inclusions have not caused a multiple increase in their wear resistance. Moreover, improving some properties due to a change in the compositions by loading with the fillers is usually accompanied by a deterioration of their other characteristics. In this regard, various optimization methods have been implemented to achieve the required properties of the polymer composites. They are often difficult to use or imply obligatory presence of a pronounced extremum of an objective function [19, 20], [etc.].

PEEK loading with fluoroplastic particles usually causes a decrease in its deformation-strength properties [13]. Absence of interfacial adhesion due to the non-polar nature of PTFE prevents formation of high-strength uniform structure. Partial loss of its strength can be compensated by loading with reinforcing fibers or improving of the polymer binder (matrix) structure (for example, by loading with (nano)fillers). As mentioned above, application of the high-modulus reinforcing fibers exerts very limited effect on metal-polymer tribological contacts. Therefore, in the present work, an attempt was made to improve the polymer matrix structure by loading with MoS₂ microparticles. This would provide solutions to several problems.

The first one was to ensure uniform structure formation during compression sintering of the polymer composite due to high thermal conductivity. The second problem was to implement dispersion hardening of the polymer, including through activation of processes at the “matrix–filler” interface. The last but not least was to provide inherent function of a solid lubricant, as complementary to action of PTFE particles. Obviously, it was difficult to increase in PEEK strength much like by loading with chopped carbon fibers, but there was a chance to reduce intense microabrasive wear of the metal counterparts. Based on the foregoing, the aim of this work was to design and optimize compositions of three-component PEEK-based composites with enhanced tribological and mechanical properties in an experimental-theoretical way that enabled to determine a range of possible filler contents.

2 Materials and Methods

The “Viktrex” PEEK powder with an average particle size of 50 μm , as well as fillers: PTFE polytetrafluoroethylene (particle size of 6...20 μm , F4-PN20 grade, “Rufflon” LLC, Russia) and MoS_2 molybdenum disulfide (Climax Molybdenum, USA, particle size of 1...7 μm) were used in these studies.

The PEEK-based composites were fabricated by hot pressing at a specific pressure of 15 MPa and a temperature of 400 °C. Subsequent cooling rate was 2 °C/min. The polymer binder powders and the fillers were mixed through dispersing the suspension components in ethanol using a “PSB-Gals 1335-05” ultrasonic cleaner (“PSB-Gals” Ultrasonic equipment center). Processing time was 3 min; generator frequency was 22 kHz.

Shore *D* hardness was determined using an “Instron 902” facility in accordance with ASTM D 2240.

Tensile properties of the PEEK-based composite samples were measured using an “Instron 5582” electromechanical testing machine. The “dog-bone” shaped samples met the requirements of Russian state standard GOST 11262-80 and ISO 178:2010.

“Pin-on-disk” dry friction wear tests of the PEEK-based composites were performed using a “CSEM CH-2000” tribometer in accordance with ASTM G99 (load was 10 N; sliding speed was 0.3 m/s). Two ball-shaped counterparts 6 mm in diameter were made of GCr15 bearing steel and Al_2O_3 ceramics (distance was 3 mm; radius of the rotation trajectory was 10 mm; rotation speed was 286 rpm). Wear rate was determined by measuring the volume of the friction track using an “Alpha-Step IQ” stylus surface profiler (KLA-Tencor, USA).

A “Neophot 2” optical microscope (Carl Zeiss, Germany) equipped with a digital camera (Canon EOS 550D, Canon Inc., Japan) was used to examine wear track surfaces after testing. Per molecular structure was studied on cleavage surfaces of notched specimens mechanically fractured after exposure in liquid nitrogen. A “LEO EVO 50” scanning electron microscope (Carl Zeiss, Germany) was used (accelerating voltage was 20 kV).

3 Results and Discussion

3.1 Two-Component “PEEK + MoS₂” Composites

Initially, two-component composites independently loaded with MoS₂ and PTFE were separately studied to evaluate effectiveness of each filler in changing PEEK mechanical and tribological properties (Sects. 3.1 and 3.2, respectively). Table 1 shows mechanical characteristics of the “PEEK + 1% MoS₂” and “PEEK + 10% MoS₂” composites (hereinafter all percentages are indicated by weight). The amount of the loaded filler was based on both published data and the results of preliminary experimental studies [10, 11, 13]. Elastic modulus increased after loading up to 10% of molybdenum disulfide particles into the polymer matrix, while tensile strength and elongation at break decreased by 11 and 54%, respectively (Fig. 1a). According to these data and taking into account an increase in Shore *D* hardness, it can be stated that loading with MoS₂ microparticles provided formation of a harder (and stiffer) composite.

SEM micrographs of the permolecular structure of neat PEEK and the PEEK-based composites are shown in Fig. 1b–d. They indicate that molybdenum disulfide was distributed quasi-uniformly mainly along the boundaries of the permolecular structure elements (Fig. 1c). The permolecular structure of neat PEEK possessed a fragmented pattern with the sizes of structural elements from units to tens of microns (Fig. 1a) which decreased after loading with 1% MoS₂ (Fig. 1b). Highly likely, finely dispersed MoS₂ particles had been crystallization centers. This effect was even more pronounced after loading with 10% MoS₂. The composite had a finely dispersed structure (Fig. 1c), most likely due to high thermal conductivity of the filler. In this case, disperse hardening (structure modification) caused an increase in hardness and elastic modulus, but, as expected, decreased elongation at break.

An increase in hardness of PEEK, modified by loading with a significant amount of filler (10%), dramatically reduced composite flexibility. Therefore, it did not contribute to improve wear resistance. The reasons were a higher material hardness and more hard wear particles, which had been formed during friction of the tribological contact parts, causing additional wear of both the polymer and the counterpart. These facts were confirmed by the results of the tribological tests presented below.

Table 1 Mechanical properties of the PEEK-based composites loaded with MoS₂

Filler composition, % (wt.)	Density ρ (g/cm ³)	Shore D hardness	Young module <i>E</i> , (MPa)	Tensile stress σ_U , (MPa)	Elongation at break ϵ , (%)
0	1.308	80.1 ± 1.17	2840 ± 273	106.9 ± 4.7	25.6 ± 7.2
+1% MoS ₂	1.310	81.1 ± 0.7	3157 ± 56	108.9 ± 2.7	12.7 ± 1.6
+10% MoS ₂	1.423	81.8 ± 0.3	3412 ± 25	96.8 ± 4.7	4.7 ± 1.4

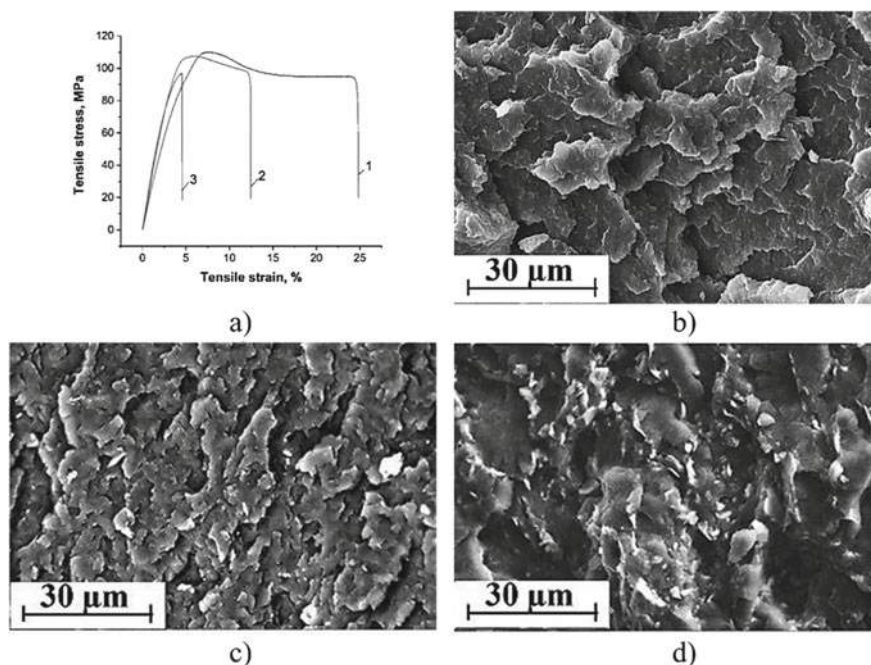


Fig. 1 Stress–strain diagram (a) and SEM-micrographs of the perfluoropolyether structure: neat PEEK (b); “PEEK + 1% MoS₂” composite (c); “PEEK + 10% MoS₂” composite (d)

Hardness of the used metal counterpart made of ball-bearing GCr15 steel was less than that of the Al₂O₃ ceramic one. In addition, the metal counterpart was able to chemically react with the polymer composite. Ceramics, in turn, was inert with respect to polymeric materials even under the conditions of tribological oxidation. As a result, chemical interaction was not supposed to occur between them. Tribological characteristics of the “PEEK + 1% MoS₂” and “PEEK + 10% MoS₂” composites are shown in Fig. 2 and Table 2. Their friction coefficient values were at the level of neat PEEK for the metal-polymer tribological contact (Fig. 2a). It is seen that MoS₂ particles in the polymer matrix did not exhibit a solid lubricant effect when slid on the softer (with respect to ceramic) metallic counterpart.

On the other hand, friction coefficient decreased by 13% in the ceramic-polymer tribological contact at a high filling degree (10% MoS₂), while at a low particle content (1%) it remained at the neat PEEK level (Fig. 2b). Thus, it was possible to realize separation of MoS₂ flakes under conditions of tougher interaction in the ceramic-polymer tribological contact, but only when filler content in PEEK was high. However, good adhesion between the polymer and the filler did not contribute to the more effective solid-lubricant action of MoS₂ particles in the two-component composite (regardless initially expected).

Wear rate of the composites increased in both metal- and ceramic-polymer tribological contacts (Table 2) despite revealed constancy or even a slight decrease in

friction coefficient values. The reasons are discussed below when analyzing wear tracks/scar on the sample and counterpart surfaces. However, the pronounced trend was a multiple increase in wear rate of the metal-polymer tribological contact compared with the ceramic-polymer one (Fig. 3). The wear rate levels were approximately the same for neat PEEK and both PEEK-based composites. These data correlated well with optical images of wear track/scar surface topography on the samples and both counterparts (Figs. 4 and 5).

Initially, the metal-polymer tribological contact was considered. According to profilometry data (Fig. 4c), wear of neat PEEK caused formation of shallow micro-grooves on the polymer friction surface (Fig. 4a). Their orientation was as usual in the sliding direction. The reason was, highly likely, micro-scratches and adherent separate debris fragments less than 200 μm in size on the metal counterpart friction surface (Fig. 4b).

PEEK loading with 1% MoS_2 solid lubricant particles caused formation of quite deep micro-grooves and scratches on the polymer friction surface (Fig. 4d and f). Surface roughness on the composite wear track was significantly greater than on neat PEEK (R_a was 0.707 μm versus 0.156 μm). Deep micro-grooves oriented along the sliding direction were also on the metal counterpart surface. Amount of

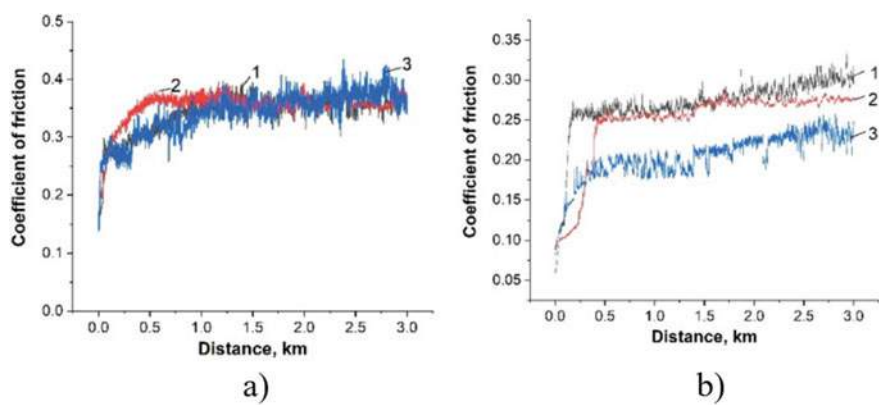


Fig. 2 Friction coefficient versus test distance: neat PEEK (1); “PEEK + 1% MoS_2 ” composite (2); “PEEK + 10% MoS_2 ” composite (3): **a**—on the metal counterpart; **b**—on the ceramic counterpart

Table 2 Tribological properties of the PEEK-based composites loaded with MoS_2

Filler composition, % (wt.)	Friction coefficient, f		Wear rate, $10^{-6} \text{ mm}^3/\text{N m}$	
	Metal counterpart	Ceramic counterpart	Metal counterpart	Ceramic counterpart
0	0.34 ± 0.03	0.27 ± 0.02	11.67 ± 1.00	3.00 ± 0.27
+1% MoS_2	0.35 ± 0.02	0.25 ± 0.02	13.67 ± 1.00	5.67 ± 0.17
+10% MoS_2	0.34 ± 0.03	0.20 ± 0.02	12.33 ± 0.33	4.00 ± 0.30

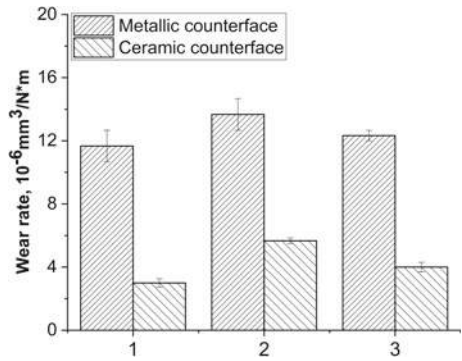


Fig. 3 Wear rate during dry sliding friction on the steel and ceramic counterparts: neat PEEK (1); “PEEK + 1% MoS₂” composite (2); “PEEK + 10% MoS₂” composite (3)

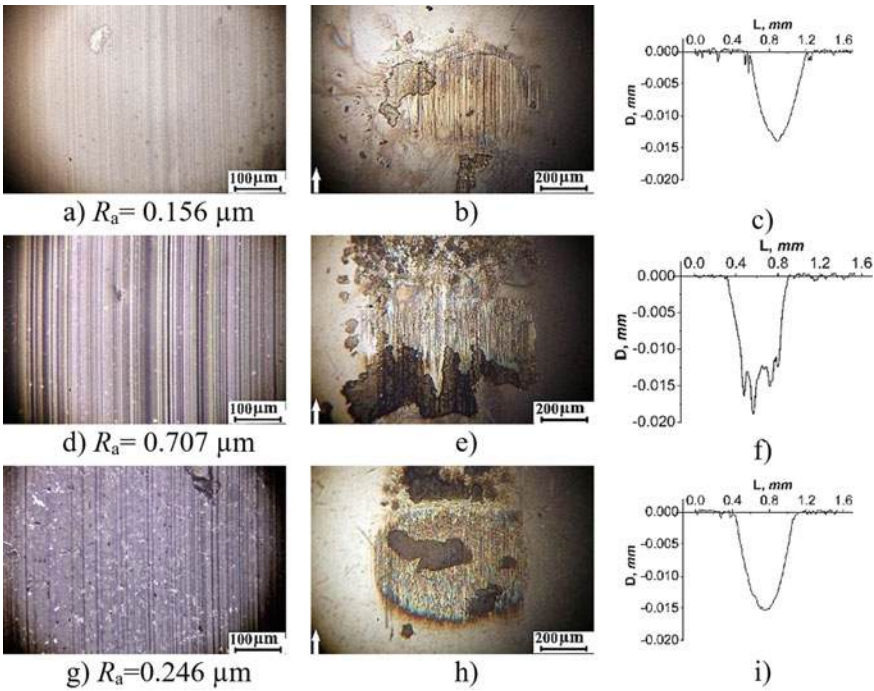


Fig. 4 Surface topography of wear scars on the samples (a, d, g), on the metal counterpart (b, e, h), and wear track profiles (c, f, i) after 3 km test distance: neat PEEK (a–c); “PEEK + 1% MoS₂” composite (d–f); “PEEK + 10% MoS₂” composite (g–i)

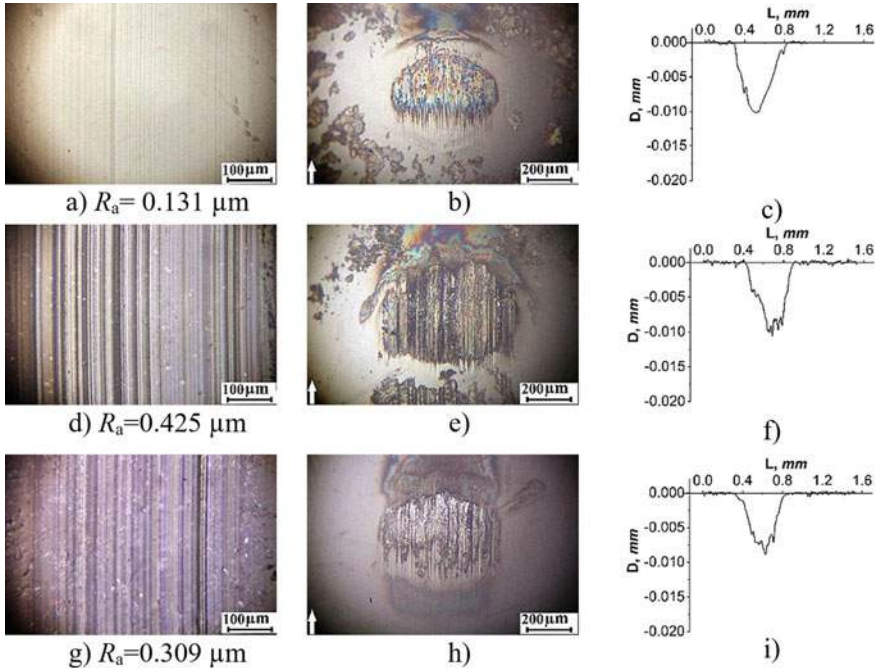


Fig. 5 Surface topography of wear scars on the samples (a, d, g), on the ceramic counterpart (b, e, h), and wear track profiles (c, f, i) after 3 km test distance: neat PEEK (a–c); “PEEK + 1% MoS₂” composite (d–f); “PEEK + 10% MoS₂” composite (g–i)

debris adhered to the metal counterpart friction surface and its wear were higher compared with neat PEEK (Fig. 4e). According to the authors, debris hardened by MoS₂ particles and oxidized during tribological loading had had a microabrasive effect on the polymer friction surface and increased its wear rate (Figs. 3 and 4d).

An increase in MoS₂ content caused composite hardness raising (Table 1). Nevertheless, the polymer composite wear track surface was smoother (Fig. 4g, i), and its roughness R_a decreased down to 0.246 μm, which was three times less than with a content of 1% MoS₂. Also, wear of the metal counterpart surface was not so heavy (Fig. 4h). Judged by the presence of rainbow colors on the surface, it can be concluded that a transfer film had been formed on it. The film had protected the metal surface from microabrasive wear by both debris and the polymer composite. Thus, after loading with 10% MoS₂, wear rate was at the level of neat PEEK despite friction coefficient was constant regardless of filling degree.

The ceramic-polymer tribological contact wear results were different. Microgrooves on the neat PEEK friction surface were also formed but their depth was much less comparing with the metal-polymer tribological contact (Figs. 4a, and 5a). However, rainbow colors were observed on the ceramic counterpart wear track (Fig. 5b) indicated that a polymer transfer film had been formed. Most likely, the

film on the surface had been the reason for a fourfold decrease in wear rate in the ceramic-polymer tribological contact (Fig. 2).

At loading 1% MoS₂, deep micro-grooves were formed on the polymer composite wear track surface (Fig. 5d). They were the same as after the tests on the metal counterpart (Fig. 4d). This fact was confirmed by contact profilometry data (Fig. 5f). At the same time, there was more intensive wear of the counterpart. This result was unexpected for hard ceramics (Fig. 5e). A transfer film was revealed on the ceramic counterpart friction surface as well.

An increase in filler content up to 10% caused a decrease in microabrasive wear both of the polymer composite (Fig. 5g) and the ceramic counterpart (Fig. 5h). At the same time, a polymer transfer film was found on the ceramic counterpart wear track. However, there were no adherent debris particles as in Fig. 5b, e. This was probably due to a decrease in composite friction coefficient.

Thus, MoS₂ molybdenum disulfide, especially when it had been slightly loaded, was not a solid lubricant for the PEEK-based composites [5, 21]. However, MoS₂ particles, due to their high thermal conductivity, had contributed to a more uniform structure formation during compression sintering. This had increased strength properties of the composite with low filler content (up to 1%). Also, MoS₂ could act as a stabilizer of fragmentary structures of multicomponent composites due to distribution of its small amount mainly on the fragment boundaries, and, thereby increasing strength characteristics. The results of such studies are presented below in the section on three-component composites.

3.2 Two-Component “PEEK + PTFE” Composites

Changes in tribological and mechanical properties of the polymer composites were different after PEEK loading with PTFE (organic) filler particles. As is known, PTFE, being solid lubricant filler, formed a transfer film on counterpart surfaces and, due to this fact, transformed tribological contacts into a polymer-polymer type [22–25]. Below are the results of studies of PEEK-based composites loaded with various amounts of PTFE chosen on the basis of both published data and previous studies of the authors [10, 12, 26].

Table 3 shows mechanical properties of the PEEK-based composites loaded with 10, 20 and 30% PTFE. Compared with neat PEEK, all mechanical characteristics of the composites decreased with increasing filler content (hardness down to 1.1 times, elastic modulus down to 1.4 times, tensile strength down to 2 times, elongation at break down to 5 times).

Despite the fact that density of the composites increased, their permolecular structures were heterogeneous: PEEK matrix elements were separated by PTFE inclusions (Fig. 6). It was expected that the higher filling degree, the less uniform structure was formed.

The results of studies of tribological properties of the PEEK-based composites loaded with various amounts of PTFE are presented in Figs. 7 and 8, as well as Table 4.

Table 3 Mechanical properties of the PEEK-based composites loaded with PTFE

Filler composition, % (wt.)	Density ρ (g/cm ³)	Shore D hardness	Young module E (MPa)	Tensile stress σ_U , (MPa)	Elongation at break ϵ (%)
0	1.308	80.1 \pm 1.17	2840 \pm 273	106.9 \pm 4.7	25.6 \pm 7.2
+10% PTFE	1.320	77.3 \pm 0.2	2620 \pm 158	83.9 \pm 2.4	5.0 \pm 0.8
+20% PTFE	1.408	75.9 \pm 0.2	2159 \pm 215	67.7 \pm 1.8	5.0 \pm 1.2
+30% PTFE	1.463	73.0 \pm 0.5	2011 \pm 108	55.1 \pm 2.1	4.7 \pm 1.4

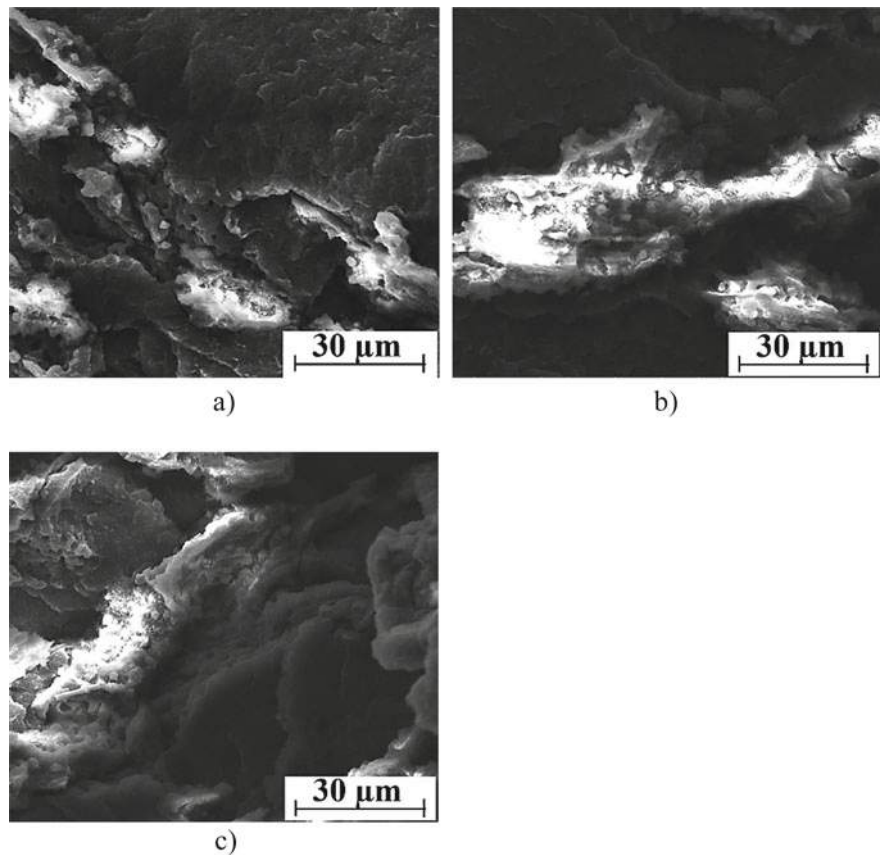


Fig. 6 SEM-micrographs of the per molecular structure of the PEEK-based composites: “PEEK + 10% PTFE” (a), “PEEK + 20% PTFE” (b), and “PEEK + 30% PTFE” (c)

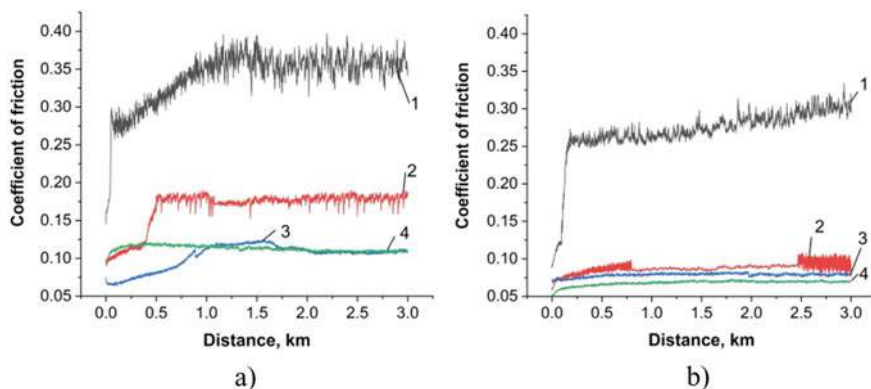


Fig. 7 Friction coefficient versus test distance: neat PEEK (1); “PEEK + 10% PTFE” composite (2); “PEEK + 20% PTFE” composite (3); “PEEK + 30% PTFE” composite (4): **a** on the metal counterpart; **b** on the ceramic counterpart

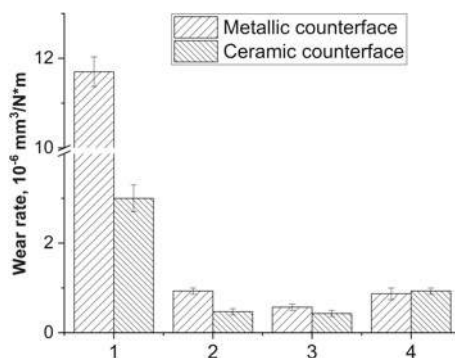


Fig. 8 Diagram of wear rate during dry sliding friction on the steel and ceramic counterparts: neat PEEK (1); “PEEK + 10% PTFE” composite (2); “PEEK + 20% PTFE” composite (3); “PEEK + 30% PTFE” composite (4)

Table 4 Tribological properties of the PEEK-based composites loaded with PTFE

Filler composition, % (wt.)	Friction coefficient, f		Wear rate, $10^{-6} \text{ mm}^3/\text{N m}$	
	Metal counterpart	Ceramic counterpart	Metal counterpart	Ceramic counterpart
0	0.34 ± 0.03	0.27 ± 0.02	11.67 ± 1.00	3.00 ± 0.27
+10% PTFE	0.17 ± 0.02	0.09 ± 0.01	0.93 ± 0.07	0.47 ± 0.07
+20% PTFE	0.10 ± 0.01	0.08 ± 0.01	0.57 ± 0.07	0.43 ± 0.07
+30% PTFE	0.11 ± 0.01	0.07 ± 0.01	0.87 ± 0.13	0.93 ± 0.07

Friction coefficient of the metal-polymer tribological contact gradually decreased by more than three times as PTFE content increased (Fig. 7a). In the ceramic-polymer tribological contact, it sharply decreased already at the minimum PTFE contents (of the studied); then it decreased slightly (Fig. 7b). This fact indicated heavier conditions of tribological loading during friction on the ceramic counterpart. As a result, PTFE inevitably acted as a solid lubricant.

Dynamics of wear resistance changes in various types of tribological contacts were significantly different. First of all, wear resistance of the “PEEK + 10% PTFE” composite increased 13.5 times in the metal-polymer and 6.5 times in the ceramic-polymer tribological contacts. However, wear rate during friction on the ceramic counterpart was two times lower in absolute terms. An increase in fluoroplastic content caused slight wear resistance rising for the “PEEK + 20% PTFE” composite. However, this improvement was not an attractive result taking into account significant deterioration of deformation-strength properties. The data from Table 4, also graphically presented in Fig. 8, enabled to conclude that the PEEK loading with 10% PTFE was sufficient to provide high wear resistance of the composites in both metal-polymer and ceramic-polymer tribological contacts.

Wear surface topographies and wear track profiles on the samples as well as counterparts' wear scars are presented and discussed below.

In the metal-polymer tribological contact, the metal counterpart was slightly worn after the “PEEK + 10% PTFE” composite test (Fig. 9b). PTFE particles were quasi-uniformly distributed in the form of rather large inclusions on the polymer composite surface (Fig. 9b) and micro-grooves almost had not been formed (Fig. 9c). On the other hand, a wear scar had been formed on the counterpart surface, whose area was smaller than that after the neat PEEK test (Figs. 4b and 9b). Also, a thin transfer film was found on the metal counterpart surface, as concluded based on the rainbow colors on the wear scar. The film, according to the authors, had protected both surfaces from (microabrasive) wear. In this case, roughness of the composite wear track surface decreased almost twofold compared with neat PEEK ($R_a = 0.081 \mu\text{m}$ versus 0.156).

The polymer composite friction surface became smoother (Fig. 9d and g) and the friction track were less pronounced (Fig. 9f, i) as filling degree increased up to 20 and 30%. However, the amount of debris rose on the metal counterpart surface, and wear track area expanded compared with the composite loaded with 10% PTFE (Fig. 9e, h). However, separate micro-scratches on the metal counterpart surface were also found for the PEEK-based composites loaded with 10 and 20% PTFE (Fig. 9b, e).

In the ceramic-polymer tribological contact, pattern of wear was generally similar (Fig. 10). The higher PEEK loading with fluoroplastic, the wider was the wear scar area on the ceramic counterpart surface (or, more precisely, not “the wear scar” but scuffs, since its wear was minimal, Fig. 10b, e, and h). Micro-grooves had not been formed on the friction surface of the polymer-polymer composites, as suggested beforehand (Fig. 10a, d and g). Furthermore, the regularity was revealed for the PEEK-based composites loaded with 20 and 30% PTFE that the higher filling degree, the more debris had been transferred onto the ceramic counterpart surface (Fig. 10b, e, and h).

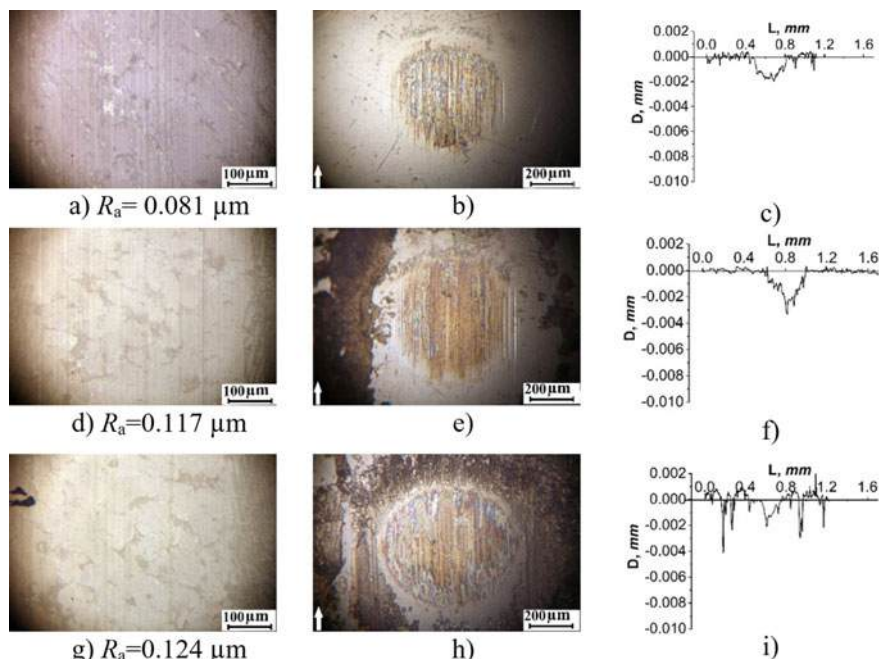


Fig. 9 Surface topography of the wear scars on the PEEK-based composites (a, d, g), on the metal counterpart (b, e, h), and wear track profiles (c, f, i) after 3 km test distance: “PEEK + 10% PTFE” (a–c); “PEEK + 20% PTFE” (d–f); and “PEEK + 30% PTFE” (g–i)

Accordingly, PTFE had formed the transfer film on the metal and ceramic counterparts, providing high wear resistance and low friction coefficient for PEEK, which in the initial state had had insufficient wear resistance for effective use in tribological contacts and a high friction coefficient of 0.34. However, PEEK loading with PTFE deteriorated structure and decreased mechanical properties. Therefore, it was suggested to additionally load with MoS_2 particles to increase mechanical and tribological properties of the “PEEK + PTFE” composites.

Presence of MoS_2 below 1% enabled to improve the process of composite formation during the sintering due to homogenization of the matrix permolecular structure. The following methodology was used to design the optimal three-component composite.

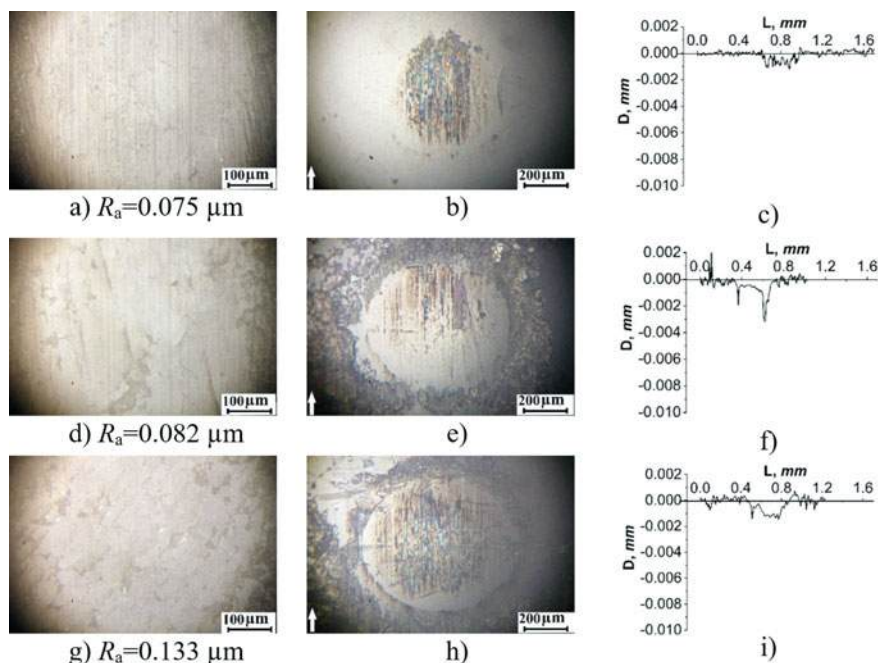


Fig. 10 Surface topography of the wear tracks on the PEEK-based composites (a, d, g), on the ceramic counterpart (b, e, h), and wear track profiles (c, f, i) after 3 km test distance: “PEEK + 10% PTFE” (a–c); “PEEK + 20% PTFE” (d–f); and “PEEK + 30% PTFE” (g–i)

3.3 Three-Component PEEK-Based Composites Filled with PTFE and MoS₂

The previously developed experimental-theoretical approach [19, 20] was used to determine the optimal composition. Twelve three-component composites were made for this purpose; their compositions are presented in Table 5. To ensure the optimal content (range of contents) of both fillers was found, the amount of PTFE was chosen to be obviously lower (5%) and higher (20%) than previously studied in its two-component composites, while the maximum MoS₂ content was 1%.

The data from the physical experiments (Tables 6, 7, 8, 9, 10, 11, 12 and 13) were used as reference points. Control parameters were PTFE and MoS₂ filling degree. When drawing surfaces for each of the control parameters, normalization was used. The lower boundary was zero; the upper boundary was unit. Additional reference points for surface drawing had been obtained using linear interpolation of the experimental data by the Lagrange polynomial [19].

Properties of the three-component PEEK-based composites were specified (Table 14) on the basis of published data and neat PEEK characteristics [27].

Table 5 Composition of the designed PEEK-based three-component composites

No.	Filler composition, % (wt.)
1	PEEK + 5% PTFE + 0.25% MoS ₂
2	PEEK + 5% PTFE + 0.50% MoS ₂
3	PEEK + 5% PTFE + 1.00% MoS ₂
4	PEEK + 10% PTFE + 0.25% MoS ₂
5	PEEK + 10% PTFE + 0.50% MoS ₂
6	PEEK + 10% PTFE + 1.00% MoS ₂
7	PEEK + 15% PTFE + 0.25% MoS ₂
8	PEEK + 15% PTFE + 0.50% MoS ₂
9	PEEK + 15% PTFE + 1.00% MoS ₂
10	PEEK + 20% PTFE + 0.25% MoS ₂
11	PEEK + 20% PTFE + 0.50% MoS ₂
12	PEEK + 20% PTFE + 1.00% MoS ₂

Table 6 Friction coefficient f of the PEEK-based composites having different MoS₂ and PTFE filling degrees during the tribological test on the metal counterpart

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	0.17	0.15	0.11	0.08
0.50% MoS ₂	0.14	0.05	0.045	0.08
1.00% MoS ₂	0.15	0.12	0.16	0.09

Table 7 Friction coefficient f of the PEEK-based composites having different degrees MoS₂ and PTFE filling degrees during the tribological test on the ceramic counterpart

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	0.14	0.08	0.06	0.047
0.50% MoS ₂	0.10	0.03	0.07	0.06
1.00% MoS ₂	0.11	0.08	0.06	0.06

Table 8 Wear rate (I , 10^{-6} mm³/N·m) of the PEEK-based composites having different MoS₂ and PTFE filling degrees during the tribological test on the metal counterpart

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	2.53 ± 0.10	1.53 ± 0.07	0.83 ± 0.10	0.60 ± 0.10
0.50% MoS ₂	2.23 ± 0.33	0.30 ± 0.03	0.33 ± 0.10	0.73 ± 0.03
1.00% MoS ₂	2.97 ± 0.33	2.67 ± 0.17	2.00 ± 0.20	0.80 ± 0.03

Table 9 Wear rate (I , 10^{-6} mm³/N·m) of the PEEK-based composites having different MoS₂ and PTFE filling degrees during the tribological test on the ceramic counterpart

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	1.00 ± 0.07	0.32 ± 0.10	0.90 ± 0.10	0.73 ± 0.10
0.50% MoS ₂	0.53 ± 0.07	0.20 ± 0.023	0.43 ± 0.07	0.43 ± 0.13
1.00% MoS ₂	1.13 ± 0.07	0.27 ± 0.07	0.37 ± 0.07	0.92 ± 0.20

Table 10 Elastic modulus (E , MPa) of the PEEK-based composites having different MoS₂ and PTFE filling degrees

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	3.08 ± 0.08	2.77 ± 0.10	2.40 ± 0.05	2.16 ± 0.07
0.50% MoS ₂	3.05 ± 0.06	2.76 ± 0.08	2.52 ± 0.07	2.07 ± 0.09
1.00% MoS ₂	2.97 ± 0.08	2.74 ± 0.04	2.73 ± 0.04	2.08 ± 0.02

Table 11 Tensile strength (σ_U , MPa) of the PEEK-based composites having different MoS₂ and PTFE filling degrees

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	94.2 ± 3.4	88.5 ± 4.7	71.2 ± 1.5	49.2 ± 2.2
0.50% MoS ₂	90.3 ± 0.7	84.9 ± 1.8	68.1 ± 1.8	45.1 ± 3.0
1.00% MoS ₂	86.5 ± 3.0	79.0 ± 0.5	66.8 ± 4.0	42.0 ± 1.4

Table 12 Elongation at break (ε , %) of the PEEK-based composites having different MoS₂ and PTFE filling degrees

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	9.9 ± 1.4	7.6 ± 0.9	7.5 ± 0.3	6.0 ± 0.6
0.50% MoS ₂	8.4 ± 0.8	9.8 ± 0.7	6.1 ± 0.3	5.2 ± 1.1
1.00% MoS ₂	7.3 ± 2.8	6.3 ± 0.4	4.3 ± 1.0	4.8 ± 0.2

Table 13 Shore D hardness of the PEEK-based composites having different MoS₂ and PTFE filling degrees

φ MoS ₂	φ PTFE			
	5% PTFE	10% PTFE	15% PTFE	20% PTFE
0.25% MoS ₂	77.9 ± 0.4	76.7 ± 0.3	76.3 ± 0.4	74.7 ± 0.5
0.50% MoS ₂	77.6 ± 0.1	77.0 ± 0.7	76.6 ± 0.5	75.5 ± 0.2
1.00% MoS ₂	78.5 ± 0.5	77.6 ± 0.3	76.9 ± 0.4	76.2 ± 0.5

Table 13.14 Specified properties for the designed three-component PEEK-based composites

Properties	Values
Shore D hardness	>75
Elastic modulus, MPa	>2500
Tensile strength, MPa	>70
Elongation at break (%)	>5
Friction coefficient on metal counterpart	<0.1
Friction coefficient on ceramic counterpart	<0.1
Wear rate on metal counterpart, 10 ^{−6} mm ³ /N m	<1.0
Wear rate on ceramic counterpart, 10 ^{−6} mm ³ /N m	<0.5

As a result, dependences of operational properties (friction coefficient, wear rate, Shore *D* hardness, elastic modulus, tensile strength, and elongation at break) on composition were obtained in the form of continuous functions. Regular data arrays reflecting the listed dependences on the control parameter discrete values were formed. Then, 3D surfaces and their corresponding contours were drawn (Figs. 11, 12, 13, 14, 15, 16, 17, 18 and 19).

The contours were overlapped to determine the values of the control parameters corresponding to the specified operational properties. The obtained range of the control parameter values, presented in Fig. 19 as a filled region, ensured that all the operational properties of the composites corresponded to the specified limits. Based on the presented data, it can be concluded that the optimal amount of MoS₂ loading was in the range from 0.4 to 0.6%, while PTFE was from 8 to 14%.

Based on the obtained data, the “PEEK + 10% PTFE + 0.5% MoS₂” composite was chosen and studied in more detail. Table 15 presents its mechanical properties and (for comparison) that of the “PEEK + 10% PTFE” one. Figure 20a shows a stress-strain diagram for these materials. The results of the analysis of these data enabled

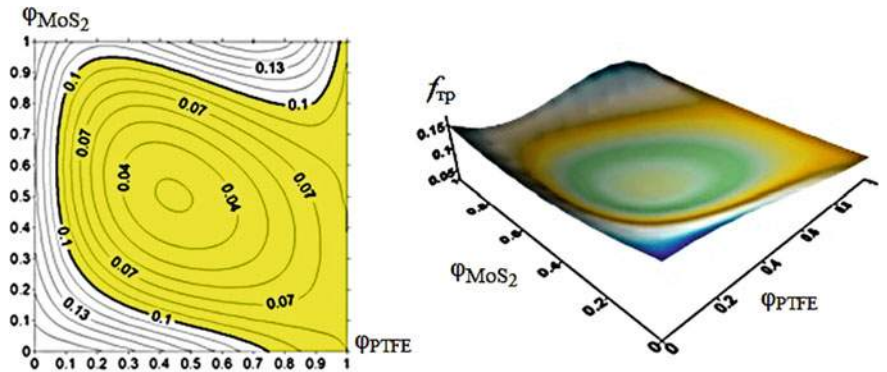


Fig. 11 Friction coefficient on the metal counterpart versus PEEK-based composite filling degree with MoS₂ and PTFE

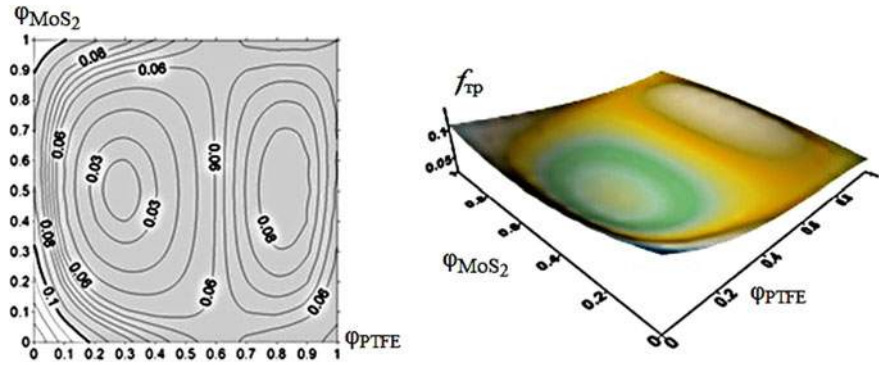


Fig. 12 Friction coefficient on the ceramic counterpart versus PEEK-based composite filling degree with MoS₂ and PTFE

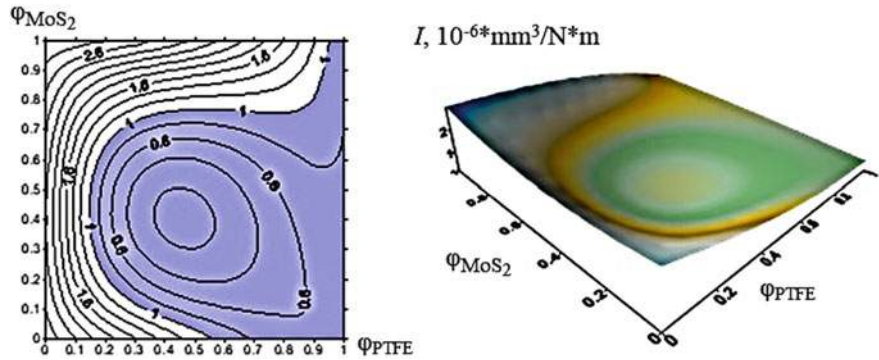
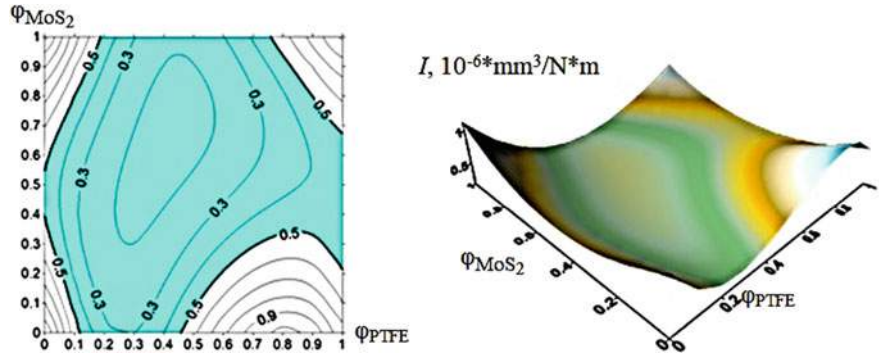


Fig. 13 Wear rate on the metal counterpart versus PEEK-based composite filling degree with MoS₂ and PTFE



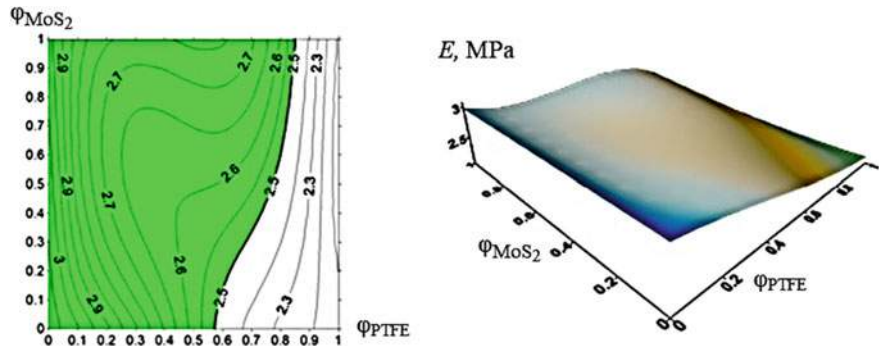


Fig. 15 Elastic modulus versus PEEK-based composite filling degree with MoS₂ and PTFE

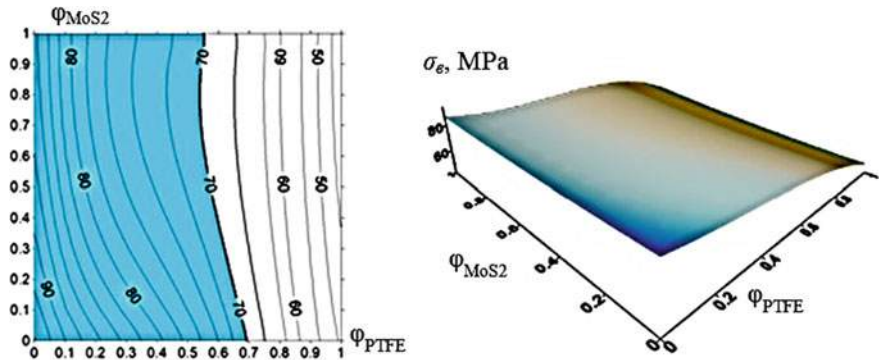


Fig. 16 Tensile strength versus PEEK-based composite filling degree with MoS₂ and PTFE

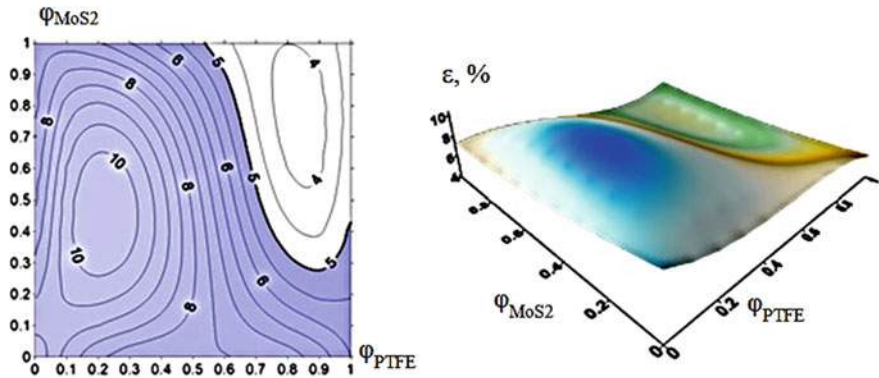


Fig. 17 Elongation at break versus PEEK-based composite filling degree with MoS₂ and PTFE

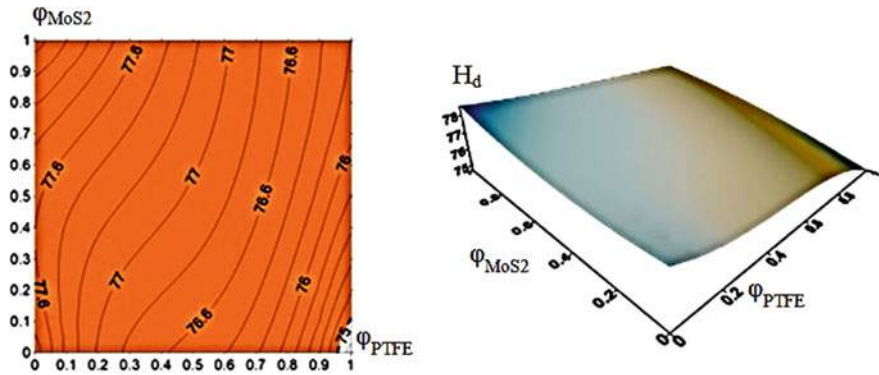


Fig. 18 Shore D hardness versus PEEK-based composite filling degree with MoS₂ and PTFE

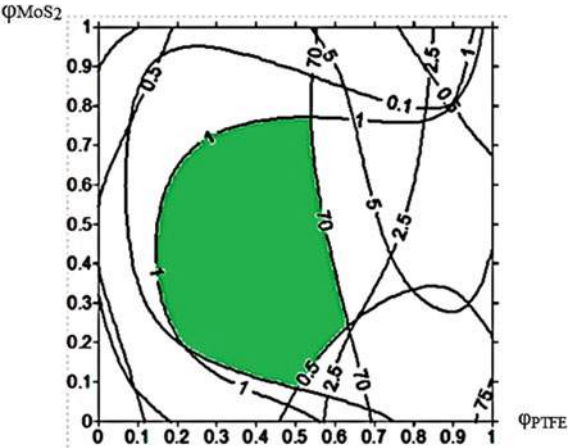


Fig. 19 Diagram of the control parameters to ensure that the mechanical properties meet the specified limits for the materials

Table 15 Mechanical properties of the optimal composition PEEK-based composite

Filler composition, % (wt.)	Density ρ (g/cm ³)	Shore D hardness	Young module E, (MPa)	Tensile stress σ_U , (MPa)	Elongation at break ε (%)
+10% PTFE (comparison)	1.324	77.3 ± 0.2	2620 ± 158	83.9 ± 2.4	4.4 ± 0.7
+ 10% PTFE + 0.50% MoS ₂	1.371	76.7 ± 0.3	2760 ± 85	84.9 ± 1.8	9.8 ± 0.2

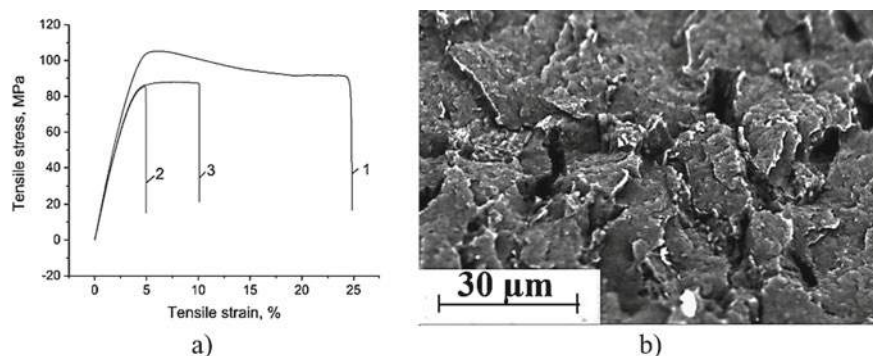


Fig. 20 Stress–strain diagram (a): 1—neat PEEK; 2—PEEK + 10% PTFE; 3—PEEK + 10% PTFE + 0.5% MoS₂; SEM-micrographs of the permolcular structure of the “PEEK + 10% PTFE + 0.5% MoS₂” composite (b)

to conclude that strength properties of the three-component composite increased slightly compared with the two-component ones. On the other hand, elongation at break doubled (Fig. 20a). Highly likely, this was due to favorable homogenization effect of 0.5% MoS₂ loaded particles on permolcular structure formation.

SEM-micrographs of the permolcular structure of the “PEEK + 10% PTFE + 0.5% MoS₂” composite are shown in Fig. 20b. It is seen that the structure was slightly loose; although there were no pronounced signs of cracking or agglomeration of each filler particles as in the case of the “PEEK + 10% PTFE” composite. According to the authors, loading with MoS₂ particles homogenized the permolcular structure due to their location along the boundaries of polymer composite structural elements. In addition to improve deformation-strength characteristics (in comparison with the “PEEK + 10% PTFE” composite), it also contributed to an increase in wear resistance. More details are discussed below.

Table 16 shows tribological characteristics of the three-component composite for dry friction on the metal and ceramic counterparts. Friction coefficient decreased by more than three times in both metal-polymer and ceramic-polymer tribological contacts. Wear resistance increased by 3.1 and 2.3 times, respectively, compared with the “PEEK + 10% PTFE” composite. Wear rate of the “PEEK + 10% PTFE + 0.5% MoS₂” composite decreased by 39 times when testing on the metal counterpart, and

Table 16 Tribological properties of the PEEK-based composites

Filler composition, % (wt.)	Friction coefficient, <i>f</i>		Wear rate, 10 ^{−6} mm ³ /N m	
	Metal counterpart	Ceramic counterpart	Metal counterpart	Ceramic counterpart
+10% PTFE	0.17 ± 0.02	0.10 ± 0.02	0.93 ± 0.07	0.47 ± 0.07
+10% PTFE + 0.50% MoS ₂	0.05 ± 0.01	0.03 ± 0.01	0.30 ± 0.03	0.20 ± 0.02

15 times on the ceramic one compared with neat PEEK. However, in absolute terms, wear rate of the three-component composite on the metal counterpart was 1.5 times higher than on the ceramic one.

Figure 21 shows friction surfaces of the samples and counterparts, as well as wear track profiles on the three-component composite. These results explain the data presented in Table 15. The counterparts almost did not wear out in both cases (Fig. 21b, e). Based on all the previously obtained data, this was most relevant for the metal one. Wear scars had been formed on both counterpart surfaces, but their area were less than that in the case of neat PEEK (Figs. 9b and 21). Micro-grooves and other damages were expectedly absent on the polymer composite friction surface, although inclusions of both fluoroplast and MoS_2 were visible.

Accumulation of a significant amount of debris in the form of a continuous film was on the metal counterpart surface, in contrast to the similar test results of the “PEEK + 10% PTFE” composite (Figs. 9b and 21b). This means that simultaneous presence of a significant content of PTFE particles and a small amount of MoS_2 in the polymer matrix had facilitated formation of a transfer film that protected the metal counterpart from microabrasive wear.

This was even more clearly shown on the ceramic counterpart surface, where the wear scar was covered with a clearly distinguishable transfer film (in the “classical” sense) which was evidenced by its rainbow reflection. The effect was most pronounced precisely in the analyzed tribological contact. It should be noted that the polymer debris clusters in the form of a uniform layer was on the ceramic counterpart surface (Fig. 21e), in contrast to the “PEEK + 10% PTFE” composite test results (as well as on the metal counterpart, Fig. 21b).

Summarizing the above, we note that a three-fold increase in wear resistance during testing both on metal and ceramic counterparts was achieved for the “PEEK

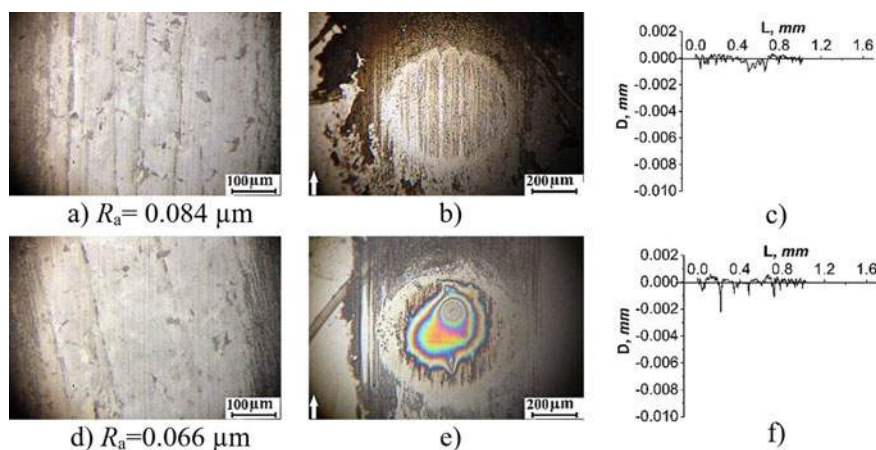


Fig. 21 Surface topography of the wear tracks on the “PEEK + 10% PTFE + 0.5% MoS_2 ” composite (a, d), on the metal (b) and ceramic (e) counterparts, and wear track profiles (c, f) after 3 km test distance

+ 10% PTFE + 0.5% MoS₂” three-component composite compared with the “PEEK + 10% PTFE”. Simultaneous loading with two types of the fillers slightly deteriorated the polymer composite structure compared with neat PEEK. However, wear rate was many times reduced due to facilitation of transfer film formation. For this reason, no microabrasive wear on both metal and ceramic counterpart surfaces developed. Besides self-lubricating effect of the three-component composite, an additional (probable) cause for metal counterpart wear eliminating was protective action of the transfer film which suppressed oxidation processes in the tribological contact of PEEK and ball-bearing steel [28].

4 Conclusions

To improve tribological and mechanical properties of polymer materials, two- and three-component PEEK-based composites loaded with molybdenum disulfide (MoS₂) and polytetrafluoroethylene (PTFE) were investigated. It was shown that an increase in dry friction mode tribological characteristics in the metal-polymer and ceramic-polymer tribological contacts was attained by loading with lubricant fluoroplastic particles. In addition, molybdenum disulfide homogenized permolecular structure and improved matrix strength properties.

A methodology for identifying composition of multicomponent PEEK-based composites having prescribed properties which based on a limited amount of experimental data was proposed and implemented. It could be used to design similar dispersion hardened composites based on prospective thermoplastic matrixes. Advantages of the methodology were shown by analysis of the experimental results on mechanical and tribological tests of the PEEK-based composites.

It was shown that wear rate of the “PEEK + 10% PTFE + 0.5% MoS₂” composite decreased by 39 times when testing on the metal counterpart, and 15 times on the ceramic one compared with neat PEEK. However, in absolute terms, wear rate of the three-component composite on the metal counterpart was 1.5 times higher than on the ceramic one. A three-fold increase in wear resistance during testing both on metal and ceramic counterparts was achieved for the “PEEK + 10% PTFE + 0.5% MoS₂” three-component composite compared with the “PEEK + 10% PTFE”. Simultaneous loading with two types of fillers slightly deteriorated the polymer composite structure compared with neat PEEK. However, wear rate was many times reduced due to facilitation of transfer film formation. For this reason, there was no microabrasive wear on both metal and ceramic counterpart surfaces. Besides self-lubricating effect of the three-component composite, an additional (possible) cause for metal counterpart wear eliminating was protective action of the transfer film, which suppressed oxidation processes in the tribological contact of PEEK and ball-bearing steel.

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Regularities of Structural Rearrangements in Single- and Bicrystals Near the Contact Zone



Konstantin P. Zolnikov, Dmitriy S. Kryzhevich,
and Aleksandr V. Korchuganov

Abstract The chapter is devoted to the analysis of the features of local structural rearrangements in nanostructured materials under shear loading and nanoindentation. The study was carried out using molecular dynamics-based computer simulation. In particular, we investigated the features of symmetric tilt grain boundary migration in bcc and fcc metals under shear loading. The main emphasis was on identifying atomic mechanisms responsible for the migration of symmetric tilt grain boundaries. We revealed that grain boundaries of this type can move with abnormally high velocities up to several hundred meters per second. The grain boundary velocity depends on the shear rate and grain boundary structure. It is important to note that the migration of grain boundary does not lead to the formation of structural defects. We showed that grain boundary moves in a pronounced jump-like manner as a result of a certain sequence of self-consistent displacements of grain boundary atomic planes and adjacent planes. The number of atomic planes involved in the migration process depends on the structure of the grain boundary. In the case of bcc vanadium, five planes participate in the migration of the $\Sigma 5(210)[001]$ grain boundary, and three planes determine the $\Sigma 5(310)[001]$ grain boundary motion. The $\Sigma 5(310)[001]$ grain boundary in fcc nickel moves as a result of rearrangements of six atomic planes. The stacking order of atomic planes participating in the grain boundary migration can change. A jump-like manner of grain boundary motion may be divided into two stages. The first stage is a long time interval of stress increase during shear loading. The grain boundary is motionless during this period and accumulates elastic strain energy. This is followed by the stage of jump-like grain boundary motion, which results in rapid stress drop. The related study was focused on understanding the atomic rearrangements responsible for the nucleation of plasticity near different crystallographic surfaces of fcc and bcc metals under nanoindentation. We showed that a wedge-shaped region, which consists of atoms with a changed symmetry of the nearest environment, is formed under the indentation of the (001) surface of the copper crystallite. Stacking faults arise in the (111) atomic planes of the contact zone under the indentation of the (011)

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surface. Their escape on the side free surface leads to a step formation. Indentation of the (111) surface is accompanied by nucleation of partial dislocations in the contact zone subsequent formation of nanotwins. The results of the nanoindentation of bcc iron bicrystal show that the grain boundary prevents the propagation of structural defects nucleated in the contact zone into the neighboring grain.

Keywords Nanocrystalline materials · Plastic deformation · Grain boundary migration · Atomic displacements · Structural defects · Shear loading · Nanoindentation · Molecular dynamics

1 Introduction

The behavior of the material in contact zones is a complex multiscale process [1], which depends on a number of parameters: the roughness of the contacting surfaces, the chemical composition of materials, loading parameters, etc. [2]. The features of fracture and wear processes in the surface layer of the materials during friction are largely determined by the shear stresses. Note that the nucleation of structural changes in materials always begins at the atomic scale. Moreover, the features of the internal structure of the material, in particular, the grain boundaries (GB), can have a significant effect on structural changes in the contact zone [3, 4]. The role of GBs in the processes of friction and wear is most significant for nanostructured metallic materials having a high GB density [5]. Note that large interest in such materials is due to their high operational properties and therefore broad prospects for their use in mechanical engineering, technology, medicine, as well as in the creation of structures for various purposes. Nanostructured metallic materials have high strength at low temperatures due to GB hardening (Hall-Petch effect). At the same time, they become superplastic at high temperatures due to GB softening, which facilitates and improves their treatment in different technological processes.

Dislocation glide is substantially suppressed in nanostructured materials [6–8]. Furthermore, the role of different modes of GB deformation or twinning is enhanced [9–12]. The main mechanism of GB deformation becomes intergranular sliding, which largely determines superplastic deformation. The experimental data and the results of computer simulations confirm the significant contribution of intergranular sliding to the plastic behavior of nanostructured metallic materials under high strain-rate loading, which lead to the formation of high local stresses [13].

GB sliding leads to the nucleation of various defects in triple junctions. These defects become sources of internal stresses and can lead to crack nucleation and brittle fracture of the material [14, 15]. The physical nature and dynamics of accommodation processes in nanostructured materials under mechanical loading are extensively studied in materials science. Typical examples of accommodation mechanisms include the emission of lattice dislocations from the zone of triple junctions, diffusion processes, rotational deformation, splitting, and migration of GBs [16]. At that, GB

migration and splitting caused by GB sliding are often realized in the form of collective self-consistent displacement of atoms in the interface region, which significantly enhances the efficiency of material accommodation on the applied loading.

A significant part of the studies on the atomic mechanisms of friction, wear, and plasticity in nanocrystalline materials is carried out using various computer simulation approaches. Still many issues related to the nucleation and development of plastic deformation, structural transformations, and wear in the contact zones in nanostructured metallic materials remain debatable. This is due to both the great variety of chemical composition and internal structure of nanostructured materials, and the difficulties in their experimental studying at the microscopic level, associated with small spatial and temporal intervals of the processes.

The considerable interest in the study of the tribological properties of nanostructured metallic materials is the elucidation of structural transformations in bicrystals with different types of GBs under mechanical loading. In [17] it was shown that some types of GBs in fcc materials can migrate with abnormally high speeds under shear loading. This can lead to a change in the structure and tribological parameters of the material in the friction zone. Therefore, the identification of mechanisms responsible for GB migration is important for the development of new approaches to stabilize the internal structure of materials in the friction zone.

The results of molecular dynamics simulation showed that the GB migration along the normal often occurs together with the tangential displacement of the grains, which leads to the shear deformation of the lattice intersected by the GB [18]. In turn, the shear stresses applied to the GB can cause its normal displacement, i.e. during GB migration, one grain will grow at the expense of another. Depending on the direction of the applied shear stresses, the GB can shift either in one direction or in the opposite direction along the normal vector.

High-rate shear loading of the crystallite can lead to the formation of vortex motion of atoms in the region of symmetric tilt GBs [19, 20]. The dimensions of the vortices in diameter are several lattice parameters and are characterized by significant atomic displacements not only in the direction of loading, but also in the GB plane. This process is dynamic, and the accommodation of the material is carried out on the basis of an abnormally high GB velocity. GB migration is based on self-consistent collective atomic displacements. Despite the fact that the displacement of each individual atom in the GB region is small, self-consistent vortex atomic displacements result in reconstruction of a significant region of one grain into the structure of the neighboring grain.

In this chapter, the regularities of structural rearrangements in the region of symmetric tilt GBs in bcc vanadium and fcc nickel initiated by high-speed shear loading are considered. Features of the behavior of materials during contact interaction are presented by studying atomic mechanisms of nucleation and development of plasticity in fcc copper and bcc vanadium metals with an ideal structure and GBs under nanoindentation.

2 Materials and Methods

Studies were carried out on the base of the molecular dynamics method using the LAMMPS software package [21]. The interatomic interaction in bcc vanadium and iron was described using many-body potentials [22, 23] calculated with the embedded atom method in the Finis-Sinclair approximation. In the case of fcc nickel and copper, the interatomic interaction was described using many-body potentials [24, 25]. Parallelepiped-shaped samples were modeled. The gamma-surface minimization algorithm [26] was used to construct GBs. Visualization of the simulation results was carried out using the OVITO package [27].

We used two parameters to identify the atoms involved in the generation of structural changes, the reduced slip vector [28] and the topological parameter that takes into account the nature of the relative positions of the nearest neighbors for each atom (common neighbor analysis) [29]. The reduced slip vector P_i is a dimensionless quantity, which is determined by the formula: where j is the nearest neighbor of atom i , N_s is the number of neighboring atoms, \vec{r}^{ij} and \vec{r}_0^{ij} are the vectors between the positions of atoms i and j in the current and initial positions, respectively.

$$P_i = \frac{1}{N_s} \sum_{i \neq j} \frac{(\vec{r}^{ij} - \vec{r}_0^{ij})}{|\vec{r}_0^{ij}|},$$

3 Features of Symmetric Tilt Grain Boundary Migration in Metals

The objects of the study were vanadium bicrystals containing about 40, 000 atoms. Periodic boundary conditions were simulated in the X and Z directions, rigid boundary conditions were set in the Y direction (Fig. 1). The initial temperature

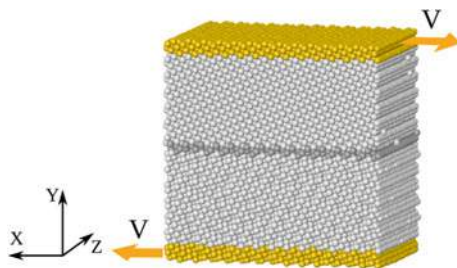


Fig. 1 The initial structure and the loading scheme of the sample containing the $\Sigma 5(210)[001]$ GB, and the loading scheme. The grips are marked in yellow, the GB region is highlighted in gray, directions of the grip shift are indicated by arrows

of the samples was 300 K. The calculations were performed for samples containing two types of symmetric tilt GBs: $\Sigma 5$ (310) [001] and $\Sigma 5$ (210) [001]. The shear loading rate in the X direction in different calculations varied from 1 to 100 m/s.

The simulation results showed that shear loading of bicrystals leads to a high-speed GB migration. We found that the velocity of GB motion is determined by the shear rate and the GB structure. The GB velocity increases with an increase in the shear rate. The average velocity of the $\Sigma 5$ (210) [001] GB for the considered shear rates is within the range from 3 to 280 m/s. The average velocity in the case of the $\Sigma 5$ (310) [001] GB is significantly lower and is in the range from 2 to 180 m/s. It is important to note that plasticity does not nucleate in the samples despite the high GB migration velocity.

During the loading, a periodic increase and drop of stresses occur. The GB migrates in a pronounced jump-like manner due to the crystallinity of the sample. Upon stress drop, the instant GB migration velocity rapidly increases, reaches a maximum value, and drops to almost zero. This is clearly seen from a comparison of the corresponding curves in Fig. 2. Note that the dependence of stresses on time in the interval of growth and drop is linear. This indicates that no structural defects are generated in the loaded sample.

Simulation results showed that atomic rearrangements in the $\Sigma 5$ (310) [001] and $\Sigma 5$ (210) [001] GB regions responsible for GB migration are significantly different. The displacement of the $\Sigma 5$ (210) [001] GB is a result of atomic rearrangements in three atomic planes: two planes of the GB and the adjacent upper grain plane. Green color shows atomic planes belonging to the GB, and blue shows the plane of the upper grain in Fig. 3 This plane adjusts to the structure of the lower grain

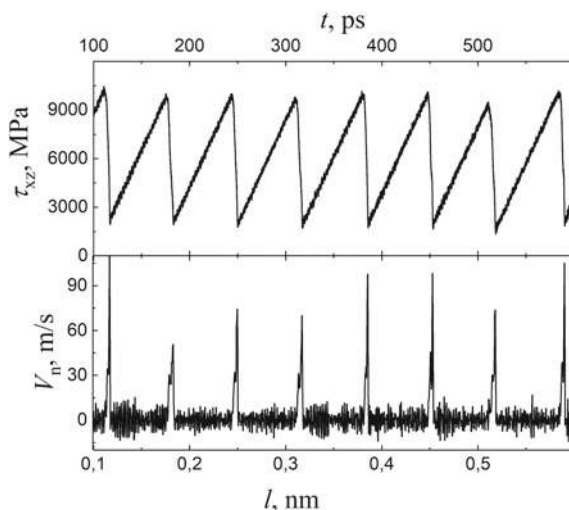


Fig. 2 The velocity of the $\Sigma 5$ (210) [001] GB migration in the normal direction (V_n), and stress τ_{xz} depending on the grip displacement. Shear rate is 1 m/s

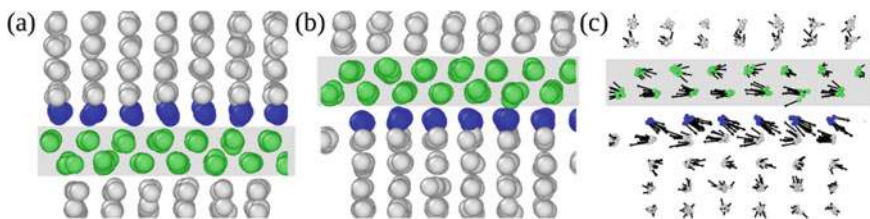


Fig. 3 Fragment of the sample with the $\Sigma 5$ (310) [001] GB before loading (a) and after the GB displacement on three interplanar distances (b). The projection of the displacements on the YZ plane (c). The shear rate is 1 m/s. The GB region is marked with gray

during shear loading (Fig. 3b). The resulting atomic displacements in the YZ plane after such a rearrangement are shown in Fig. 3c. The value of these displacements is about 0.08 nm. Analysis of simulation results showed that the incorporation of the atomic plane of the upper grain into the structure of the lower grain is realized as a sequence of three successive displacements in different directions. The duration of each displacement for the grip velocity of 1 m/s is about 3.6 ps. The values of these three displacements are approximately equal to 0.07, 0.03, and 0.06 nm. As a result of such displacements, the atoms of the blue plane successively occupy the positions of the atoms of the upper and then the lower GB planes and, finally, adjust to the lattice of the lower grain.

Three atomic planes are simultaneously involved in the displacement of the $\Sigma 5$ (210)[001] GB. The four atomic planes make up the GB (highlighted in gray), and the three atomic planes (highlighted in blue, green, and red) belong to the upper grain in Fig. 4. As a result of a certain sequence of displacements the upper grain planes first rearranged into the structure of GB planes and then adjust to the structure of the lower grain. The value of the resulting displacements of the atoms of the blue plane during its transformation to the lower grain is 0.05 nm. During this interval, two jump-like displacements occur, each with a duration of 4.0 ps. The value of the first displacement is 0.04 nm, the second is 0.05 nm. After the first displacement, the

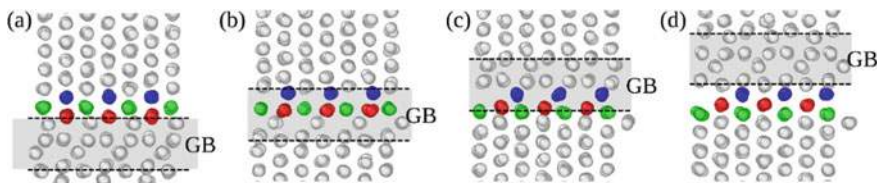


Fig. 4 Fragment of the structure with the $\Sigma 5$ (210)[001] GB before loading (a), structural configuration of colored planes after first (b) and second (c) GB displacement, and after the GB was displaced on four interplanar distances (d). The GB region is located between the dashed lines and is highlighted in gray. The shear rate is 1 m/s

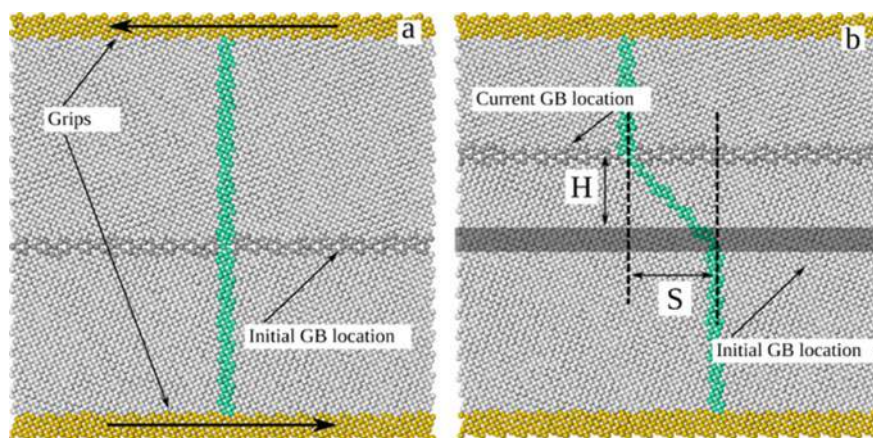


Fig. 5 Crystallite structure and the GB position at different points in time after the start of loading: 0 ps (a); 300 ps (b). Arrows in a show the displacement direction of the grips. The green line shows the position of the atoms forming a vertical line along the middle of the sample before loading

selected planes are rearranged into the GB structure (Fig. 4b). The second displacement changes the stacking order of green and red planes and then adjusts the green plane to the lower grain (Fig. 4c, d).

Note that the atomic planes far from the GB also have a pronounced periodicity of motion in the normal direction. It is due to the crystallinity of the sample structure and the constant velocity of the grips. However, the jump-like motion of these planes is less pronounced in comparison with atomic planes belonging to GB.

The structural transformations leading to high-speed migration of GBs in fcc metals under shear loading were studied for nickel bicrystals containing the $\Sigma 5$ (310)[001] symmetric tilt GB. The simulated sample was composed of about 70,000 atoms. Periodic boundary conditions were simulated in directions parallel to GB, and rigid boundary conditions were set in the third one. The loading scheme and the position of the GB are shown in Fig. 5a. The initial temperature of the sample was 300 K. The shear loading rate in different simulations varied from 1 to 100 m/s.

The simulation results showed that shear loading of nickel sample causes a high-speed GB motion along the normal to its plane. To analyze the peculiarities of the GB motion in the bicrystal, a vertical layer was selected with a thickness of several lattice parameters normal to the GB plane. It was found that the atoms of this layer in the interval between the initial and final positions of the GBs have a pronounced displacement gradient (Fig. 5b). Atoms outside this interval are displaced by equal distances with grips. This is a shear-induced displacement of the GB. Such character of the GB displacement was revealed experimentally in [30]. It is one of the main mechanisms of grain growth and is quite common in the processes of recrystallization of the structure. For a quantitative description of this GB displacement, a coupling factor is introduced. It is defined as the ratio of the lateral (S) and normal (H) displacement values:

$$b = S/H.$$

The coupling factor depends on the structure of the GB region and the sample temperature [18]. For simulated GB, the coefficient b is approximately equal to 1.

The change in the shear stress τ_{xy} during the GB motion is shown in Fig. 6. One can see that the GB moves in a jump-like manner, which is due to the sawtooth nature of the change in shear stress. Note that the segments of the curve on which the shear stress increases correspond to the flat segments on the GB displacement curve. The segments of the τ_{xy} curve at which the drop occurs correspond to the GB displacement. The pronounced periodicity of the curves in Fig. 6 is associated with the discreteness of the crystallite structure and the symmetry of the GB. At the same time, thermal fluctuations of the atomic system cause deviations from this periodicity.

Analysis of the simulation results showed that the GB motion is realized through a certain sequence of transformations of typical structural elements in the intergranular region. Figure 7 shows a fragment of the structure in the GB region at different points in time. This fragment contains two atomic planes along the normal to the plane of the figure. The Roman numerals denote the atoms in the GB region, which belonged to the upper grain at the initial moment. Note that the atoms with numbers I and III belong to the same plane in the normal direction to the figure, and II and IV belong to the other plane. We revealed that the transformation of the GB atomic layer of the upper grain into the structure of the lower grain occurs through three characteristic jump-like displacements of numbered atoms in the XY plane. The directions and values of these three displacements for numbered atoms are shown in Table 1. The GB motion in the lateral and normal direction to its plane during loading is always provided by the indicated displacements of the boundary atoms.

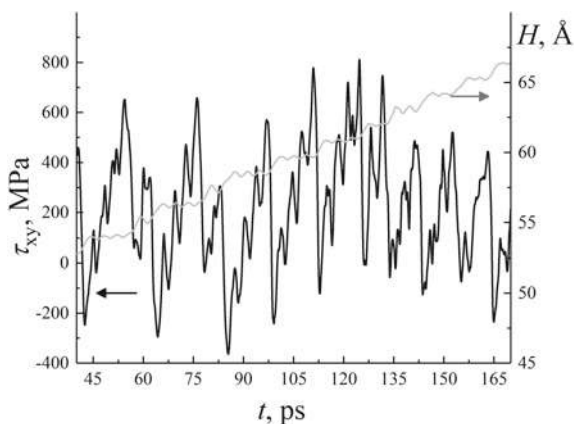


Fig. 6 Dependence of the shear stress and coordinates of the $\Sigma 5$ (310)[001] GB position in nickel bicrystal on time. The black curve is shear stress; the gray curve is the GB position

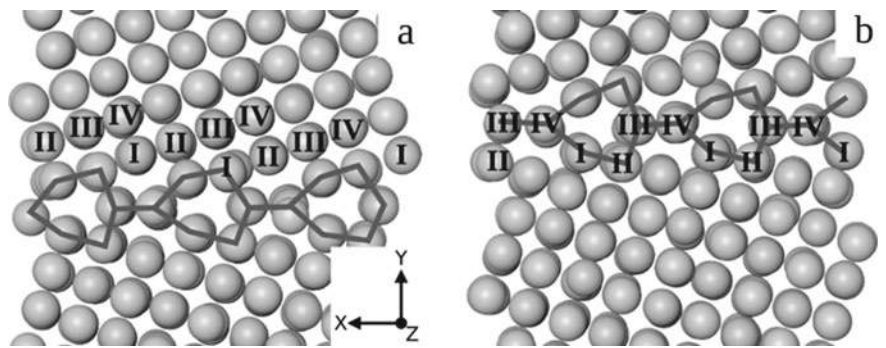


Fig. 7 Fragment of the structure containing the $\Sigma 5$ (310)[001] GB in nickel at the following points in time after the start of the shear: **a** 22.5 ps; **b** 43 ps. Lines indicate the GB structural elements

Table 1 Atomic displacements providing GB motion under shear loading

Atom numbers	First displacement, Å (X; Y)	Second displacement, Å (X; Y)	Third displacement, Å (X; Y)
I	(0.7; 0.1)	(0.4; 0.0)	(−0.5; 0.0)
II	(0.6; 0.0)	(0.3; −0.5)	(−0.4; −0.4)
III	(0.5; 0.0)	(0.5; 0.1)	(0.6; 0.0)
IV	(0.6; 0.0)	(0.4; 0.0)	(0.4; −0.4)

Note that the motion of the symmetric tilt GB is realized without nucleation of structural defects. The use of periodic boundary conditions does not allow grain rotation during the displacement process. Since a tilt symmetric boundary is simulated, both grains have the same shear moduli in the direction of applied loading. Therefore, the GB motion is completely due to the coupling effect.

4 Peculiarities of Plasticity Nucleation in Metals Under Nanoindentation

One of the most informative and effective methods for studying the physical and mechanical properties of materials during contact interaction is nanoindentation. A change in the indentation conditions allows a systematic study of the influence of various factors on the processes occurring in the contact zones of the materials. As a rule, the aim of works related to computer simulation of the material behavior under indentation is to study the mechanisms of plastic deformation in the zone of a spherical or pyramidal indenter, visualize defect structures, and interpret load-indentation curves [31–34]. Despite the high information content of such studies, it is difficult to analyze the results due to the complex deformation pattern. For the

clearer and simpler interpretation of the indentation results, it is convenient to use an extended indenter of a cylindrical shape [35–37]. For this choice of the indenter, the contact region is linearly extended from one face of the sample to the other. The loading scheme of a copper crystallite with such an indenter is shown in Fig. 8. The axis of the indenter was parallel to the loaded crystallite surface. Free boundary conditions were set along this axis. The loaded face was a free surface, while atomic positions of several layers of the opposite face were fixed in the indentation direction. The lateral faces of crystallites were simulated as free surfaces. The indentation rate was 25 m/s. The simulated crystallites were loaded at 300 K.

To study the behavior of simulated crystallite under indentation the loading force was calculated as a function of the indentation depth. The loading force (F) was defined as the total force acting on the indenter from the loaded crystallite. The indentation depth (d) was calculated as the distance from the lower boundary of the indenter surface to the level of the crystallite surface in the initial state. The results of the calculation of the loading force - indentation depth dependence are shown in Fig. 9.

The indenter and crystallite begin to interact as soon as the distance between them becomes smaller than the cutoff radius of the interatomic interaction. Initially, an attractive force arises between the indenter and the loaded surface (this corresponds to a negative value of the loading force in Fig. 9). This effect is called “jump-to-contact” [38]. During the indenter displacement, the attractive force changes into repulsive. In accordance with this, the loading force in Fig. 9 has a pronounced minimum. The dependence of the indentation force on the indentation depth can be divided into four stages. The first stage is characterized by a linear dependence of the loading force on the indentation depth and corresponds to the elastic response of the material. At the beginning of the second stage, local structural transformations are generated in the contact zone (Fig. 10a). The generation of such local structural transformations

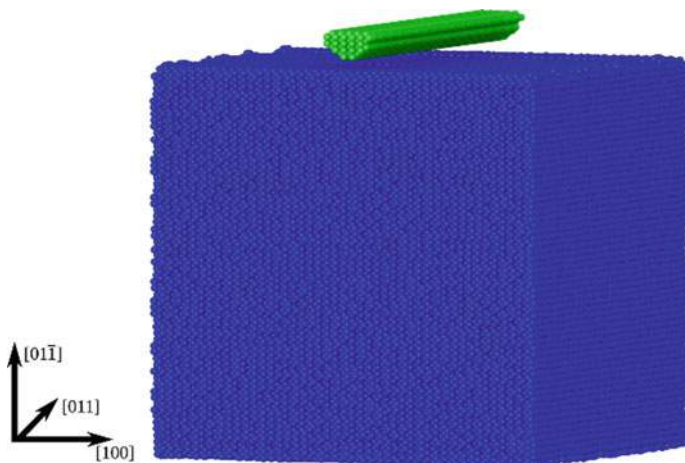


Fig. 8 Indentation scheme and crystallographic orientation of the simulated system

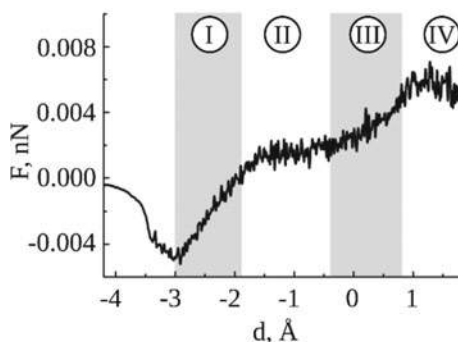


Fig. 9 Dependence of the loading force on the indentation depth

in an ideal crystallite leads to a partial relaxation of excess stresses and a decrease in the slope of the curve with its subsequent transition to plateau in Fig. 9. The change in the behavior of the loading force during the transition from the first stage to the second is also associated with the discreteness of the indenter structure. In particular, during the indenter penetration, new atomic layers of the indenter start to interact with the free surface of the crystallite. The number of defects in the contact zone at the second stage quickly reaches saturation, and their number at the third stage changes insignificantly. This is due to the fact that the mechanism of excess stress relaxation by the generation of structural defects exhausts itself, resulting in an increase in the slope of the loading curve. An analysis of the simulation results shows that further indenter penetration (the beginning of the fourth stage) leads to an intensive increase in the number of local structural transformations. This leads not only to a slowdown in the growth of the loading force but to its superseeding by a decrease. Moreover, local structural transformations lead to the nucleation and development of structural defects of a higher rank, in particular, intrinsic and extrinsic stacking faults (SFs) (Fig. 10b). Structural defects in the contact zone are generated in the $\{111\}$ atomic planes. Formed defects can spread along the indicated planes to the free surfaces. Their escape to the free surface leads to the formation of steps that change the crystallite shape. The defect structure in the loaded crystallite at the moment of the SF escape to the free surface is shown in Fig. 10c. The step on the free surface can be seen in Fig. 10d.

Note that the crystallite structure, its crystallographic orientation, indenter shape, loading scheme, and boundary conditions have a significant effect on the simulation results. To study the effect of the orientation of the loaded surface on the response of the material during indentation, the calculations were carried out for three surfaces with small Miller indices: $(01\bar{1})$, (001) and (111) . The simulated samples had the shape of a cube with a side of 165 \AA and consisted of 350,000 atoms. The loading scheme was similar to that described earlier, except that periodic boundary conditions were set along the axis of the cylindrical indenter. The indentation speed was 50 m/s . The influence of the orientation of the loaded surface on the formation of structural defects was studied based on the calculation of the dependences of the indentation

force $F(d)$ and the fraction of atoms n involved in local structural changes on the indentation depth d (Fig. 9).

The calculation results of the $F(d)$ and $n(d)$ curves for the $(01\bar{1})$ orientation of the loaded surface are shown in Fig. 11. Marks on the $F(d)$ curve indicate the indentation depths at which the loading curve abruptly changes the angle of inclination or has pronounced kinks. Projections of the crystallite structure for indentation depths corresponding to these points are shown in Fig. 12. We found that the features of the $F(d)$ and $n(d)$ curves correlate well with each other. In particular, the calculations show that the local structural changes lead to a decrease in the slope of the indentation force curve or to the appearance of an extremum in the loading curve due to relaxation of internal stresses.

Initially, an attractive force begins to act on indenter as it approaches the sample (the region of negative values in Fig. 11), and reaches its absolute maximum at a distance of 3.0 Å from the loaded surface. This “dip” in the curve of the loading force is accompanied by a bending of the loaded surface towards the indenter.

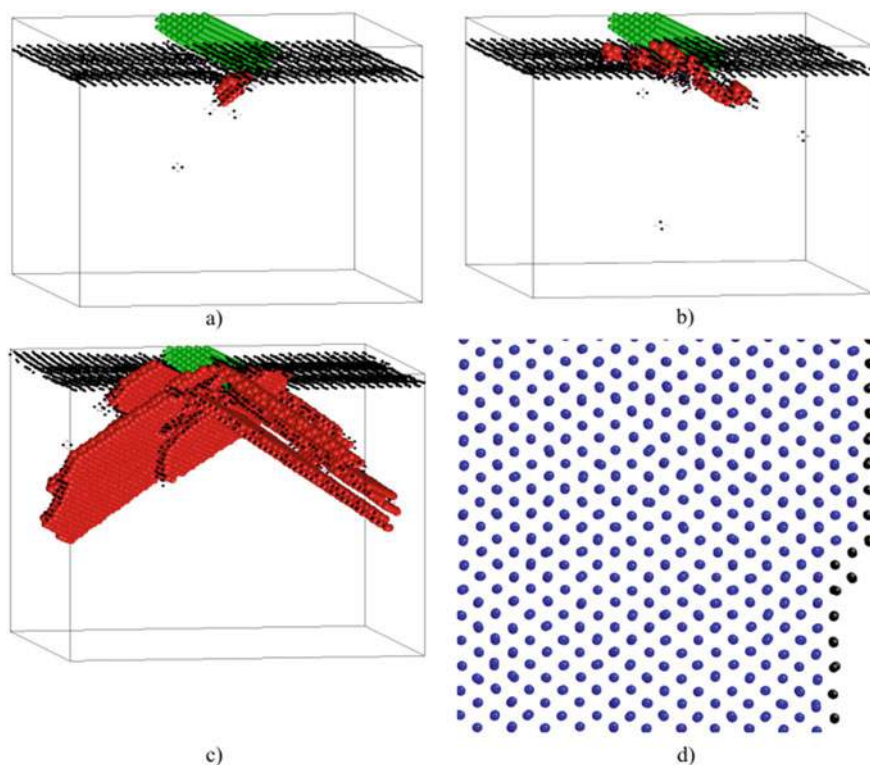


Fig. 10 Fragment of the simulated crystallite for indentation depths: **a** -1.0 Å, **b** 1.5 Å, **c** 8.7 Å. Atoms with the fcc symmetry of the nearest environment are not shown. Green spheres indicate indenter atoms, red large spheres, and dark points indicate atoms with hcp and undefined symmetry of the nearest environment, correspondingly. **d** a step on the free surface

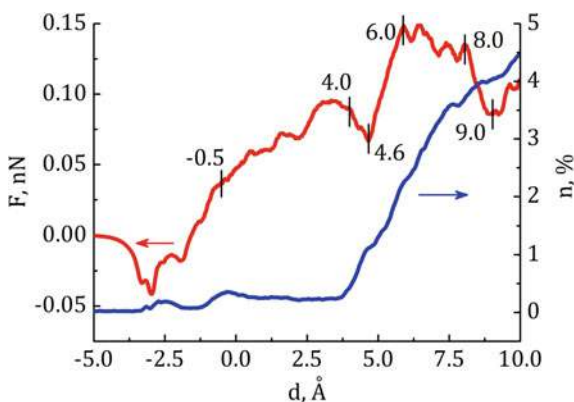


Fig. 11 Dependences of the loading force (red curve) and the fraction of atoms involved in local structural changes (blue curve), on the indentation depth in the sample with the (01 $\bar{1}$) loaded surface

A change in the shape of the loaded surface leads to the generation of local structural changes in the region under the indenter. An analysis of the structure showed that the atoms in this region have 12 nearest neighbors, but their environment does not correspond to any of the known lattices. With further indentation, the surface returns to its previous position, the number of defects decreases, and the loading force remains practically unchanged when the indenter moves in the range from -2.5 to -1.8 Å. Then the loading force begins to grow with a constant number of defects. After passing the depth of -0.5 Å, its slope decreases. This is due to an increase in the number of local structural changes that form $a/6 < 112 > \{111\}$ partial dislocations, and then SFs. They are located in adjacent $(11\bar{1})$ and $(\bar{1}11)$ planes (Fig. 12a). Further, the number of local structural changes slightly decreases. In this case, dislocations remain motionless, which leads to a further increase in the loading force.

Starting from an indentation depth of 3.6 Å, new SFs are generated and grow in the crystallite. They are located in the area under the indenter in planes of the same type, but deeper than previously formed SFs. The latter also begin to increase in size (Fig. 12b). In this case, a significant decrease in the loading force is observed, which continues until the indenter depth reaches 4.6 Å. The number of defects continues to increase at further loading, but the loading force begins to grow. This is due to the fact that the trailing partial dislocations move from the contact zone towards the free surface (Fig. 12c, d). At an indentation depth of 6.0 Å, the maximum value of the loading force is reached. The escape of dislocations to free surfaces leads to change of the shape of the crystallite, in particular, to the formation of steps on the free surface (Fig. 12e, f).

The simulation results of the copper crystallite behavior upon indentation of the (001) free surface are shown in Fig. 13. The peculiarities of the loading force curve in this figure correlate well with the peculiarities of the curve describing the number of atoms involved in local structural changes.

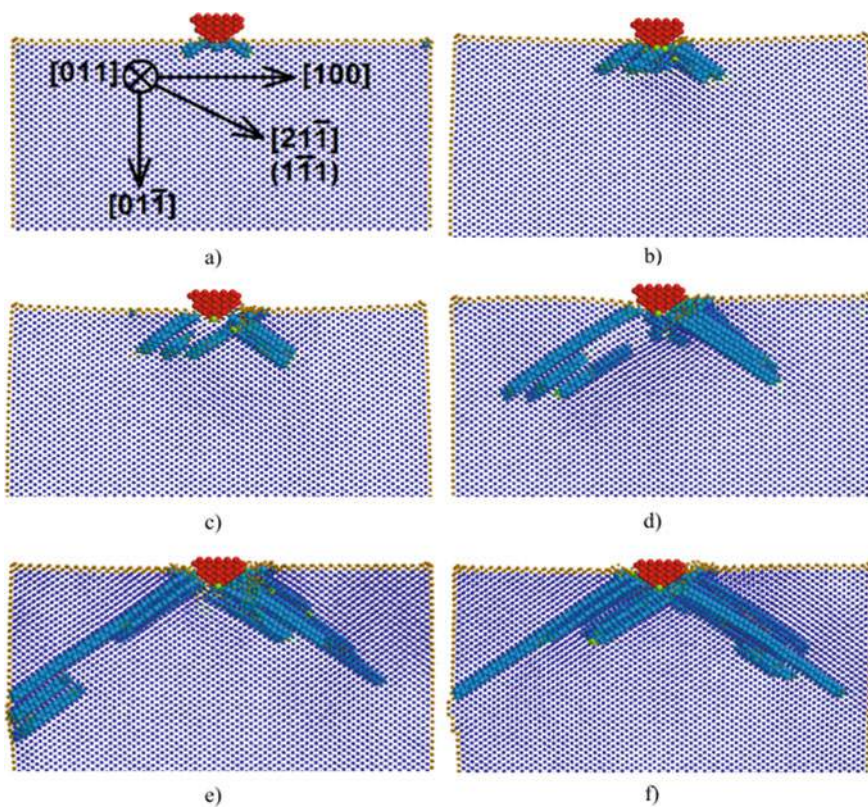


Fig. 12 Fragment of the crystallite structure with the $(01\bar{1})$ loaded surface at different indentation depths: **a** -0.5 \AA , **b** 4.0 \AA , **c** 4.6 \AA , **d** 6.0 \AA , **e** 8.0 \AA , **f** 9.0 \AA . Large blue and green spheres show the atoms with hcp and undefined symmetry of the nearest environment, respectively. Indenter atoms are marked in red

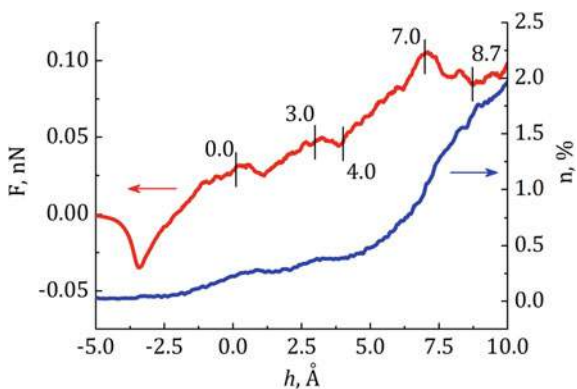


Fig. 13 Dependences of the loading force (red curve) and the fraction of atoms involved in local structural changes (blue curve) on the indentation depth in the sample with the (001) loaded surface

The attractive force of the indenter to the surface reaches its maximum value at a distance of $\sim 3.4 \text{ \AA}$. The number of defects increases due to the bending of the free surface. At further indentation, the loading force and the number of atoms involved in local structural changes begin to grow. Note that the curve of the loading force has features at the indentation depths of 0 and 3 \AA . They are related to the discrete structure of the indenter. As the indenter penetrates the material, the atomic rows of the indenter alternately interact with the crystallite. Since distant atomic rows are initially attracted to the loaded surface, this leads to a slowdown in the growth of the loading force.

An analysis of the structure of the simulated crystallite showed that at the indentation depth of 0.8 \AA , a region containing atoms with either hcp or undefined symmetry of the nearest environment is formed in the contact zone (Fig. 14a). The sizes of this region are comparable with the sizes of the indenter. With further indentation, the width of the region along the $[010]$ direction doubles and then does not change. At the same time, the size of the defective region along the $[001]$ direction increases with the penetration depth of the indenter. Note that a more intensive increase in the density of defects leads to a decrease in the loading force of the sample at an indentation depth of 7.0 \AA .

Due to the inertia of accommodation processes, structural changes in the crystallite continue for some time after the indenter stops. During relaxation after the indenter stops at a depth of 10.0 \AA , dislocation loops continue to move from the top of the region containing defects to the side free surfaces of the sample (Fig. 14c–e). The defect escape on free surfaces leads to the formation of steps (Fig. 14f), and the fraction of local structural changes decreases from 2.0 to 0.3%.

The simulation results of the copper crystallite behavior under indentation of the (111) free surface are shown in Fig. 15. The curves presented in this figure correlate quite well with each other.

Calculations showed that local structural changes begin to form in the sample at the indentation depth of -1.7 \AA (Fig. 16a). Their formation slows down the growth of the loading force. The SF starts to grow in the $(11\bar{1})$ free surface at the indentation depth of 0.8 \AA (Fig. 16b), and the loading force changes slightly.

The leading and trailing $a/6 < 112 > \{111\}$ partial dislocations are generated and move in adjacent $(11\bar{1})$ planes during loading of the crystallite (Fig. 16c–e). As a result of this process, a twin is formed in the crystallite. Atoms with hcp symmetry of the nearest environment are located on its boundaries (Fig. 16f). Note the formation of a fragmented region in the contact zone. This region consists of hcp atoms and grows towards the right side free surface (Fig. 16e, f).

In the case of bcc iron, the indenter was modeled by a repulsive force field in the form of a cylinder. The use of such an indenter reduces the effect of the structure discreteness on the structural response of the material. The sample dimensions with an ideal structure were $170 \times 170 \times 170 \text{ \AA}$ (Fig. 17a), and the dimensions of sample the with a GB was $230 \times 170 \times 170 \text{ \AA}$ (Fig. 17b). The projection of the structure of the $\Sigma 13 (320)[001]$ symmetric tilt GB is shown in Fig. 18. The axis of the cylindrical indenter was oriented parallel to the loaded surface of the crystallites. The indenter force field was described by the formula: where R is the indenter radius, r is the

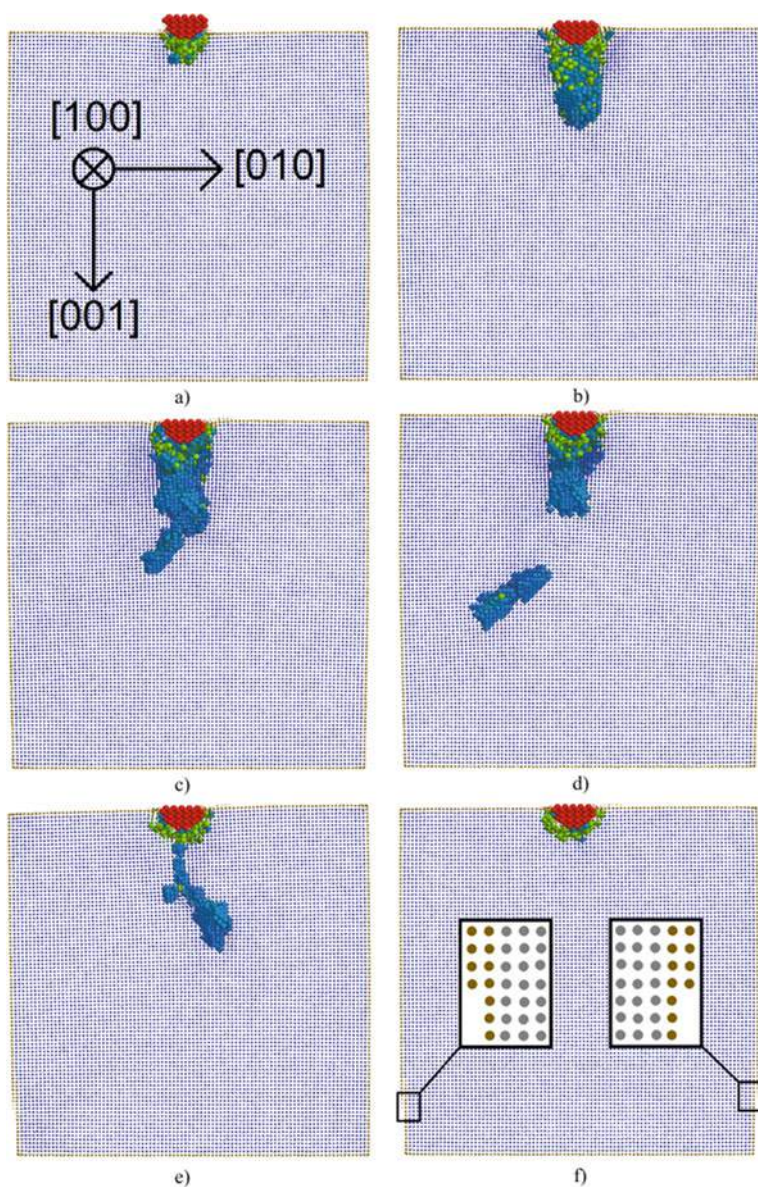


Fig. 14 Crystallite structure under indentation of the (001) free surface at different indentation depths: **a** 4.0 Å, **b** 8.7 Å. The crystallite structure at different points in time after the indenter was stopped at the depth of 10 Å: **c** 40 ps, **d** 60 ps, **e** 80 ps, **f** 120 ps. Large blue and green spheres show the atoms with hcp and undefined symmetry of the nearest environment, respectively. Indenter atoms are marked in red

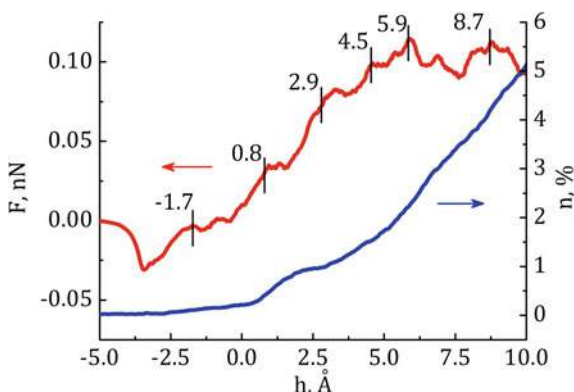


Fig. 15 Dependences of the loading force (red curve) and the fraction of atoms involved in local structural changes (blue curve), on the indentation depth in the sample with the (111) loaded surface

distance from the indenter axis to the atom. The loading scheme was similar to that described previously for fcc samples of different orientations. The indenter was pressed in at a constant speed of 1 m/s. The kinetic temperature of the samples was 300 K.

$$U = \begin{cases} -\frac{(R-r)^4}{4}, & r < R \\ 0, & r > R \end{cases},$$

The dependences of the loading force on the indentation depth are shown in Fig. 19. The loading force curves for both crystallites are similar. The regions of linear growth of the loading force correspond to the elastic response of crystallites. The onset of plastic deformation (the formation of local structural changes) can be determined by an abrupt decrease in the loading force. An analysis of the indentation results showed that atoms involved in local structural changes have the value of the reduced slip vector exceeding 0.2. Note that, for a crystallite with the GB, plastic deformation nucleates at smaller indentation depths. Such a response is associated not only with the GB presence in the crystallite but also with the fact that the indenter contact line is oriented differently with respect to the loaded surface than in the case of a crystallite with an ideal lattice. The growth and drop of the loading force in Fig. 19 correlate well with curves showing a change in the number of atoms forming local structural changes.

An analysis of the simulation results shows that local structural changes initially nucleate in the region of contact of the indenter with the surface, and then propagate along the slip planes towards the lateral faces of the crystallite (Fig. 20). Their escape to the side face leads to the formation of a step on the free surface. Note that the GB prevents the propagation of local structural changes in the neighboring grain, which is clearly seen in Fig. 20b.

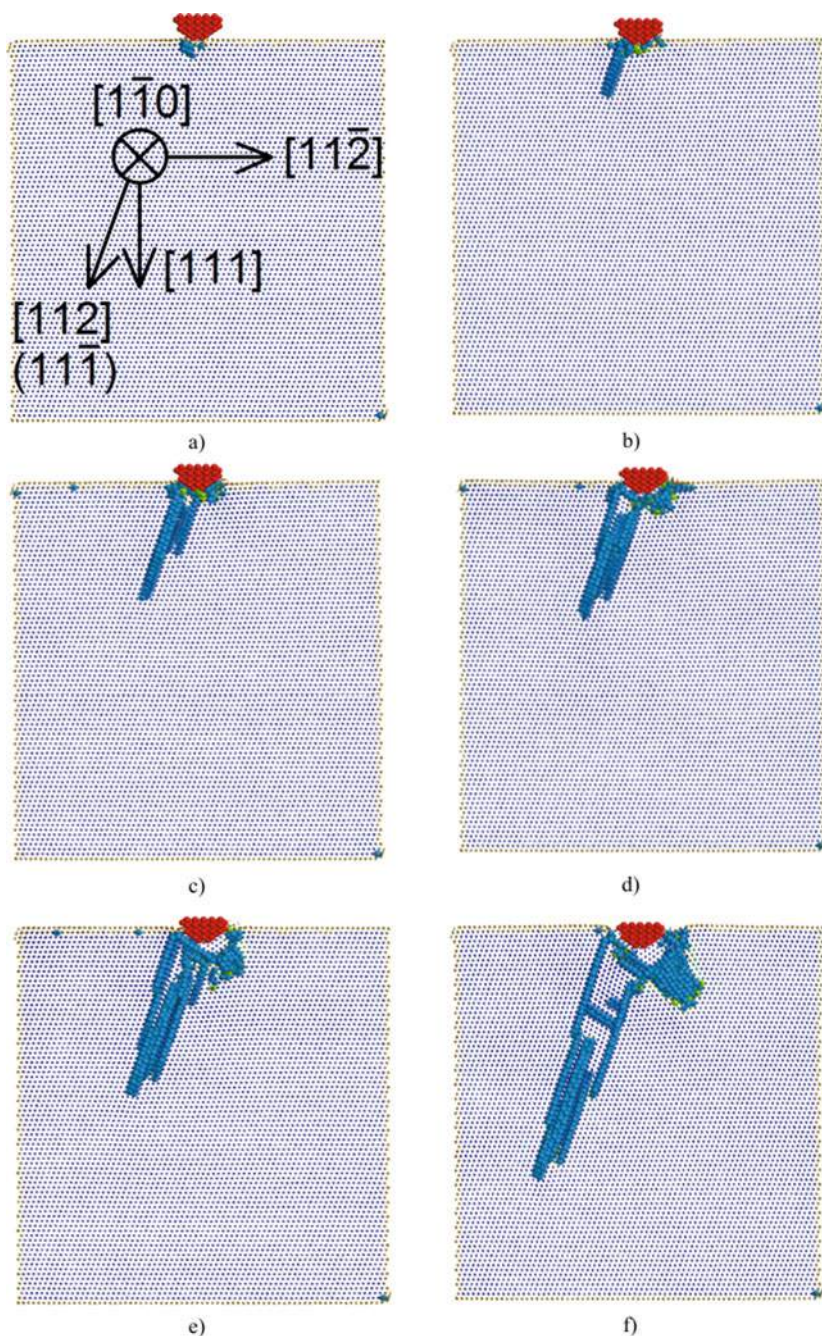


Fig. 16 Crystallite structure under indentation of the (111) free surface at different indentation depths: **a** – 1.7 Å, **b** 0.8 Å, **c** 2.9 Å, **d** 4.5 Å, **e** 5.9 Å, **f** 8.7 Å. Large blue and green spheres show atoms with hcp and undefined symmetry of the nearest environment, respectively. Indenter atoms are marked in red

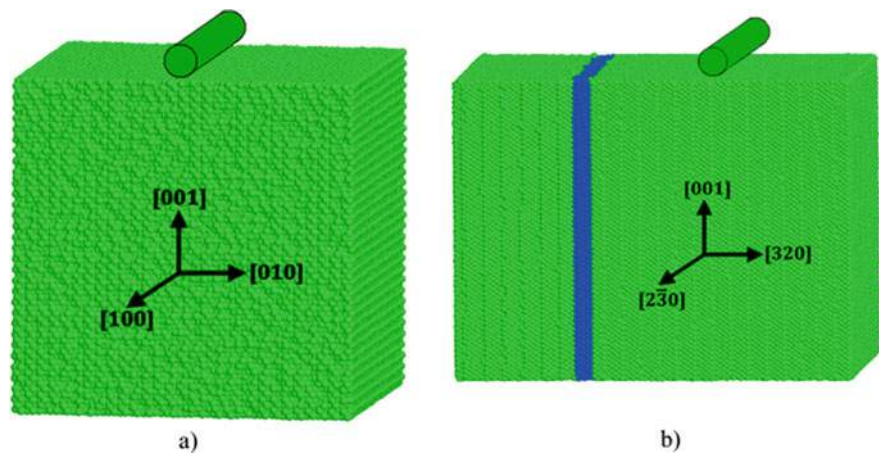


Fig. 17 Scheme of loading for the sample with an ideal lattice (a) and with the GB (b). The GB region is marked in blue

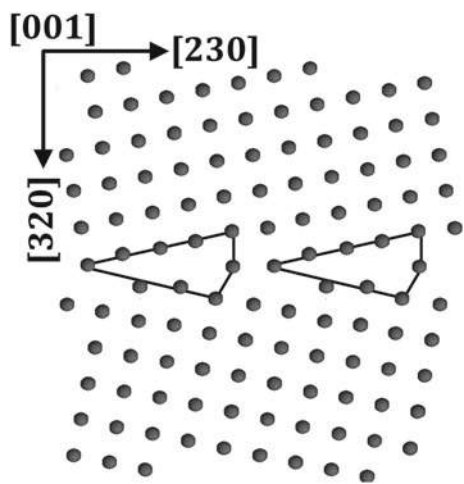


Fig. 18 Projection of the $\Sigma 13$ (320)[001] symmetric tilt GB structure onto the (001) plane after relaxation. Solid lines indicate GB structural elements

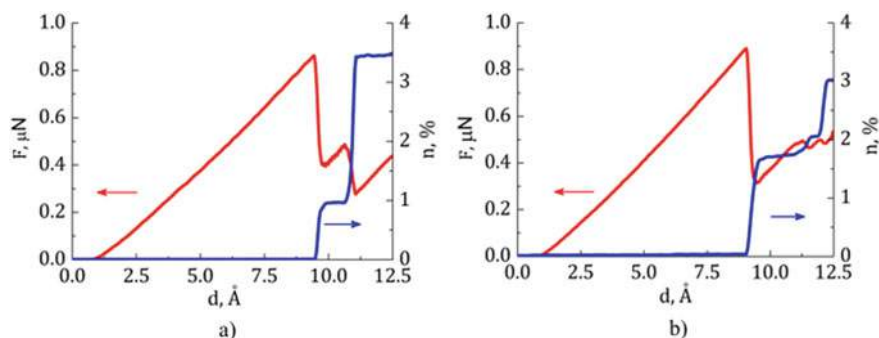


Fig. 19 Loading force (F) and the fraction of structural defects (n) depending on the indentation depth (d) for the single crystal (a) and the sample with the GB (b)

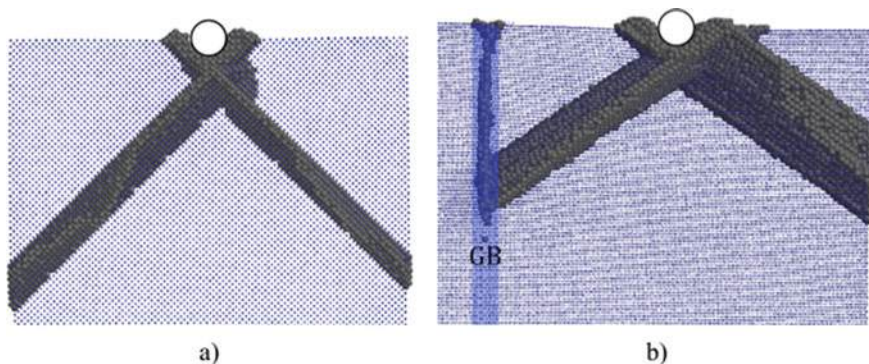


Fig. 20 Projections of the sample structure with an ideal structure (a) and with the GB (b) for an indentation depth of 12.5 \AA . Dark gray spheres show atoms which reduced slip vector is greater than 0.2

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Fault Sliding Modes—Governing, Evolution and Transformation



Gevorg G. Kocharyan, Alexey A. Ostapchuk, and Dmitry V. Pavlov

Abstract A brief summary of fundamental results obtained in the IDG RAS on the mechanics of sliding along faults and fractures is presented. Conditions of emergence of different sliding regimes, and regularities of their evolution were investigated in the laboratory, as well as in numerical and field experiments. All possible sliding regimes were realized in the laboratory, from creep to dynamic failure. Experiments on triggering the contact zone have demonstrated that even a weak external disturbance can cause failure of a “prepared” contact. It was experimentally proven that even small variations of the percentage of materials exhibiting velocity strengthening and velocity weakening in the fault principal slip zone may result in a significant variation of the share of seismic energy radiated during a fault slip event. The obtained results lead to the conclusion that the radiation efficiency of an earthquake and the fault slip mode are governed by the ratio of two parameters—the rate of decrease of resistance to shear along the fault and the shear stiffness of the enclosing massif. The ideas developed were used to determine the principal possibility to artificially transform the sliding regime of a section of a fault into a slow deformation mode with a low share of seismic wave radiation.

Keywords Fault · Earthquake · Slow slip event · Stick-slip · Seismic waves · Rock friction

1 Introduction

One of the important areas of Professor S. G. Psakhie’s activities was studying regularities of deformation of hierarchical blocky media. His interest in these problems arose, among other things, under the influence of Academician S. V. Goldin. Sergey Grigorievich took active part in his seminar. Special attention in his works was paid to the possibility of altering the regime of blocky medium deformation to prevent powerful seismic impacts [1–3]. Starting with the application of the method

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of movable cellular automata [4] to the problems of deformation of contact areas between rock blocks, S. G. Psakhie and his colleagues conducted a series of experimental works, including investigations of the reaction of a natural fault to dynamic disturbances and watering [5–7]. Especially a big cycle of works should be noted that had been performed on the ice cover of the Lake of Baikal, which was used as a model to study the tectonic processes in the Earth's crust [8, 9].

Authors of this paper were lucky to take part in experiments that were held under the guidance of S. G. Psakhie at a segment of the Angarskiy seismically active fault (Baikal Rift zone) [10]. The device constructed in IDG RAS—the Borehole Generator of Seismic Waves (BGSW, Fig. 1)—was mounted there to disturb the fault. The device was used to produce periodical explosions of the air-fuel mixture in a specially drilled borehole [11]. The results of observations showed that after the active disturbance the nature of movements along the fault changed. Precise measurements of the parameters of displacement along the fault showed that the disturbance triggered movements along strike, which manifested as a left-side shear. The quasi-dynamic micro-displacements which were detected many times in the



Fig. 1 The Borehole Generator of Seismic Waves (BGSW) mounted at one of the sides of the Angarsky fault

records after actions with a drop-hammer, explosions and BGSW coincided with the macro-displacement in direction. Approximately in a day after the action of BGSW had been terminated the background direction of the creep and the average velocity of sliding ($\sim 4 \div 5 \mu\text{m/day}$) restored [12].

Approaches that were developed in those works had a clear physical sense. Injecting water and the effect of vibrations can both change the parameters of friction during the shear along the fault and increase the pore pressure. These changes may result in emergence of conditions that correspond to the Mohr–Coulomb failure criterion. It means that a movement can be provoked at a fault that hasn't reached the ultimate state yet. Though the mechanics of the process seemed evident, it remained unclear whether the triggered movement would be a “quasi-viscous” one (in terms of V. V. Ruzhich with co-authors [7]) and no dynamic failure would occur. Let us cite Sh. Mukhamediev: “Here the clarity in fault's behavior ends. Further development of the artificially triggered movement (velocity of its propagation and its size) is not so evident. Here the non-uniformities of properties and stresses along the existing fault or its future trajectory play the most important role” [13]. The macroscopic conditions of different sliding modes on faults remained uncertain at that time. For the last 10–15 years the geophysical community has essentially advanced in many components of fault mechanics—cumulating information about the structure of segments where slip localizes [14–16], conditions of dynamic slip to be initiated [17, 18], in situ observations [19, 20] and laboratory modeling of slip episodes on faults [21–23]. We are going to present some of these data in this chapter.

2 Fault Slip Modes

In the very beginning of instrumental observations over deformations of the Earth's surface it became clear that stresses cumulated in tectonically active regions relax not only through dynamic failure of some sections of the Earth's crust, but through continuous aseismic sliding (creep) along existing faults, too. Earthquakes were interpreted as a quasi-brittle failure of rock, while creep—as a plastic deformation. It was believed that in the areas where the rate of deformation is high enough, accumulation of elastic stresses occurs with further dynamic failure of rock accompanied by intensive emission of seismic waves. In case the rate of deformation of a limited volume of the medium is so low that stresses have time to relax on all the structural inhomogeneities, regimes of deformation at a constant velocity without destruction (creep) occur [24]. Thus, it was believed that the earthquake and the aseismic sliding are two opposite phenomena that take place under different loading conditions in the medium.

As the observation data were cumulated and measuring facilities were upgraded, qualitative and quantitative differences between seismic events of one and the same rank were detected. For example, it turned out that seismic energies emitted in earthquakes with approximately equal seismic moments can differ by several orders of magnitude [25, 26].

Sensitive deformographs and tiltmeters periodically registered displacements and deformations at velocities several orders of magnitude higher than the background ones, but essentially slower than the velocity of rupture propagation in an “ordinary” earthquake. However, the low installation density of such devices didn’t allow to summarize the data being obtained, especially as the attention of investigators was concentrated primarily on post-seismic and pre-seismic deformations.

The situation changed qualitatively when dense networks of GPS sensors and broadband sensitive seismic stations were launched to operation in a continuous regime [27, 28]. As a result, in the last 25–30 years the fault slip modes were detected and classified as what can be treated as transitional from the stable sliding (creep) to the dynamic failure (earthquake). Discovering these phenomena changed to a great extent the understanding of how the energy cumulated during the Earth’s crust deformation releases—slow slippage along faults are apprehended not as a special sort of deformation, but they span a continuum of slip modes from creep to earthquake [29].

Studying the conditions of occurrence and evolution of transitional slip modes can give new important information about structure and laws of fault behavior. That is why investigations of these “unusual” movements on faults have become one of the leading trends. Detecting the phenomenon of episodic tremor and slip (ETS) in many subduction zones is believed to be one of the most important advances of geophysics in recent history [30].

Developing observation systems has allowed to reveal a number of new deformation phenomena associated with discontinuities of the Earth’s crust—subduction zones [31], continental fault zones [32], tectonic fractures [10], fractures in large ice masses [33] and even with micro-cracks in hydrocarbon reservoirs [34].

The classification of deformation events along tectonic discontinuities adopted currently is based primarily on duration of the process in the source [29, 35]. Only several percent of the released deformation energy are emitted in the form of seismic waves in a ‘normal’ earthquake (duration of the process in the source 0.1–100 s). This turns out to be enough for the strongest macroscopic manifestations of powerful earthquakes to occur. The ratio of the emitted seismic energy E_s to the seismic moment M_0 varies in the range of $E_s/M_0 \sim 10^{-6}$ – 10^{-3} , the average value being $\sim 2 \times 10^{-5}$ [25, 26].

Under some conditions, the slip velocity may not reach seismic slip rates, but nevertheless, low amplitude low-frequency seismic waves are emitted. These are the so called Low Frequency Earthquakes (LFE) and Very Low Frequency Earthquakes (VLFE) [29]. The spectrum of these vibrations is depleted with high frequencies which testifies a longer (than it follows from standard relations) duration of the process in the source—up to hundreds of seconds. The ratio of the emitted seismic energy to seismic moment, specific for LFEs, is about $E_s/M_0 \sim 5 \times 10^{-8}$ – 5×10^{-7} , and the velocity of rupture propagation is $V_r \sim 100$ – 500 m/s [36–38]. VLFEs have durations in the source of about tens to hundreds of seconds, a velocity of rupture propagation of about $V^r \sim 10$ – 100 m/s and an energy-to-moment ratio of $E_s/M_0 \sim 10^{-9}$ – 10^{-7} [39].

In some cases, the peak slip velocity is so low that seismic waves which could be recorded instrumentally are not emitted at all. Nevertheless, the slip velocity in these deformation phenomena noticeably exceeds typical velocities of aseismic creep along faults which is, on average, several centimeters a year. Such deformation events that can last from several hours to several years are called Slow Slip Events (SSE).

First reports in which the phenomena of aseismic sliding were for the first time interpreted as “slow earthquakes” came after observations at the Izu Peninsula in Japan [40]. But perhaps for the first time, an episode of slow slip as a self-contained event that had a start and a termination was described by A. T. Linde with co-authors [32]. The authors presented a recorded deformation event about a week in duration. They called it a “slow earthquake” and proposed to characterize such events quantitatively, just like ordinary earthquakes, with the value of seismic moment M_0 or moment magnitude M_w , which is linked to the seismic moment by the well known relation [41]:

$$M_w = 2/3(\lg M_0 - 9.1) \quad (1)$$

The velocity of rupture propagation along the fault strike for SSEs lies in the range from several hundreds of meters a day to 20–30 km a day [39]. Despite small displacements, an appreciable seismic moment accumulates at the expense of a large fault area, at which the displacements take place. An essential part of energy cumulated in the course of deformation releases through slow movements. For example, in New Zealand about 40% of seismic moment release through SSEs [42]. Slow displacements are registered there with durations from several months to a year and with moment magnitudes up to $M_w \sim 7$. Such SSEs repeat with a period of 5 years. Less scale events with durations of several weeks have a recurrent time of 1–2 years [43, 44].

Large scale SSEs last for months and even years. The seismic moment released in them is comparable to the one of the most powerful earthquakes. For example, from 1995 to 2007 more than 15 events were registered in different regions around the world, each of them with the released seismic moment of more than $M_0 \sim 5 \times 10^{19}$ Nm, which corresponds to the moment magnitude of $M_w \sim 7$ [39]. Their durations were from a month to one year and a half, and the amplitude of displacement along the fault reached 300 cm. It should be noted that some of these SSEs were not independent events, but episodes of post-seismic sliding.

The results of studying slow movements along faults show that these specific deformations are widespread all around the world and to a great extent at their expense stress conditions of many segments of the Earth’s crust are regulated. Though initially it was thought that periodical slow slip is specific mainly for depths of several tens of kilometers in the subduction zones, installation of dense networks of seismic and geodetic observations allowed to detect similar phenomena at shallow sections of submerging plates and continental faults [45]. It is not impossible, that as the density and sensitivity of installed devices increase, sections of periodic slow slips with small moment magnitudes will be detected at numerous tectonic structures, including areas

of high anthropogenic activities. Slope phenomena have also much in common with slow tectonic slip along faults [46].

The importance of studying the slow slip modes bases on several reasons. Investigating mechanisms and driving forces of these processes will allow to advance essentially in understanding regularities of interactions of blocks in the Earth's crust, and, consequently, in assessing risks of natural and man-caused catastrophes linked to movements along interfaces—earthquakes, fault-slip rock bursts, landslides, etc.

Slow movements along faults can, beyond all doubt, be triggers of dynamic events. Detecting transitional slip modes all around the world has led many investigators into the attempt of linking the slow slip phenomena to powerful earthquake triggering [47–50], but the mechanics of this process is not developed so far. There are only few works, in which sequences of deformation events of different modes were registered instrumentally, and their interrelations were soundly demonstrated [51, 52]. Manifestations of seismicity were registered most reliably after slow slip [53], or manifestations of seismicity in the form of non-volcanic tremor against the background slow slip [54]. Geodetic and seismic observations can give only a confined insight into the physical mechanism of slow sliding. For example, though it is admitted that an aperiodic slip preceded the Tohoku 2011 earthquake [55], it remains uncertain, how and why the sliding regime altered just before the main shock. Taking measurements near the surface, it is actually impossible to detect small spots of “accelerated slip” at a seismogenic depth [56].

Last but not least, a problem which regularly attracts the attention of the scientific community shall be mentioned—the possibility to alter the seismic regime of some area or the deformation regime of a specific fault segment through some external actions [57]. Therefore, it is important to investigate the conditions under which different slip modes emerge and evolve in fault zones.

3 Localization of Deformations and Hierarchy of Faults

Applying the ideas of self-similar blocky structure of Earth's crust [58] inevitably leads to the necessity of introducing a hierarchy of interblock gaps—tectonic fractures and faults. Meanwhile the situation seems unobvious for these objects. At first glance, it is hard to reveal a similarity between a closed crack in a rock mass and a large fault zone, as opposed to the rock blocks they bound.

Unlike blocks whose linear sizes can be reliably measured, it is often impossible to estimate unambiguously the geometric characteristics of discontinuities. Analysis of any tectonic map shows that prolonged linear structures can be considered to be single objects only partially. Each of these structures is a concatenation of separate sections adjoining distinctly detectable blocks of certain ranks. Widening of the zones of active faults, as well as delta-wise and trapezium-wise sections of fault splitting can be observed in tectonic junctions.

The size of an active zone can be limited to a local section, even though a rather long linear structure may be available. As a rule, sizes of active zones match well

with the sizes of structural blocks, which corresponds to the classical concept that the energy of an earthquake is controlled by a certain size of a block being unloaded. Actually, it means that the length of a linear structure should be characterized by the length of its active section, which in its turn manifests quasi-independently in the geodynamic sense. The hierarchical rank of a fault zone is determined by the rank of the blocks it separates.

Despite many publications, the relationship between parameters of fault zones such as length, width (the size across strike), amplitude of displacement are being discussed actively. Empirical scale relations linking fault length L , fault width W and amplitude of displacement along rupture are widely used in describing structural characteristics of fault zones [59, 60]. Power relations of the following types are often used to establish links between these parameters:

$$W = \alpha \cdot D^a, \quad D = \beta \cdot L^b, \quad W = \chi \cdot L^c \quad (2)$$

In many publications the indexes in Eqs. (2) are more often close to one, while the factors α , β and χ vary in a wide range. Some authors expressed essential doubts in applicability of Eqs. (2). The doubts based mainly on a noticeable dispersion of experimental data [61, 62]. Closeness of indexes in Eqs. (2) to one means fulfillment of similarity relations for the process of faulting—all the linear sizes are linked to each other through direct proportionalities.

More detailed investigations of the last years [63, 64] have shown that there are several hierarchical levels, in which alterations of parameters of the events with scale occur according to different laws, which differ very often strongly from the similarity relations. Figure 2a presents maximum displacement along a fault versus fault length. The plot uses the data of several investigations.

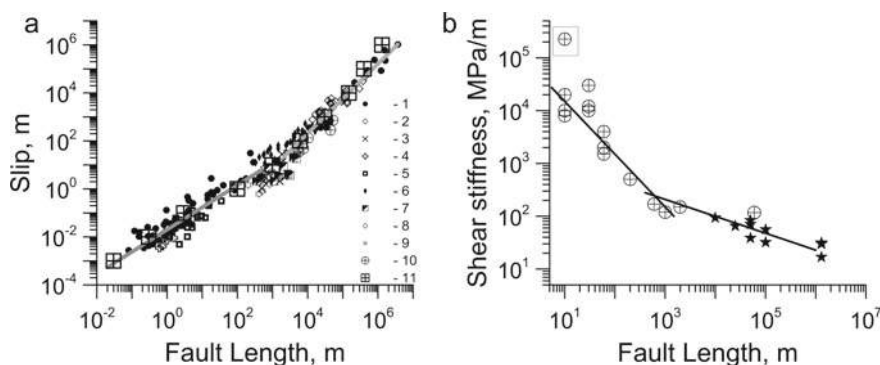


Fig. 2 Structural and mechanical characteristics of faults versus fault length. **a** Cumulative slip along a fault. References [1–11] are presented in [65]. Lines—best fit of the data in the range $L < 500$ m and $L > 500$ m. **b** Shear stiffness of a fault. Crosses in circles—the data were obtained with the method of seismic illumination. The point in the frame was not used in constructing the regression dependence. Black stars—the data were obtained with the method of trapped waves [65]

By all appearances the linear size $L \sim 500\text{--}1000$ m is a transitional zone, a specific boundary between two bands, in which the scale relations differ. Faults that have reached this stage of evolution may be called the “mature” ones. In [63] for example, a relation linking the width of the influence zone to the fault length was suggested:

$$W = \begin{cases} 0.15 \cdot L^{0.63}, & L \leq 500 \div 1000 \text{ m} \\ 0.85 \cdot L^{0.42}, & L > 500 \div 1000 \text{ m} \end{cases} \quad (3)$$

Here, the width of the influence zone is the width of the cross-section with higher fracturing.

It is important to emphasize that the change of mechanical characteristics of faults (fault stiffness) with scale demonstrates the presence of approximately the same transitional zone (Fig. 2b).

Further alteration of scaling relations is observed for the most powerful deformation events with characteristic sizes exceeding the thickness of the crust.

Investigations performed for the last 20–30 years have allowed to essentially widen and clarify the knowledge about the inner structure of fault zones. New data have been acquired in the frames of the international program of “fast drilling” of a fault zone after an earthquake [66]. Similar projects of drilling through fault zones have already been fulfilled in several regions [67, 68]. Together with the results of traditional geologic explorations at the surface and in mines [14–16, 69, 70] these data allow to acquire a rather orderly notion about the structure of faults’ central parts.

The *damage zone* is located at the periphery of the fault. Its width may vary from meters to hundreds of meters. It is usually associated with a higher fracture density, if compared to the intact massif. The damage zone contains distributed fractures of a wide range of sizes. It is structured to a great extent and usually contains a standard set of discontinuity types [71].

Cataclastic metamorphism becomes more intensive in the direction to the *fault core*. One or several sub-zones of intensive deformations can usually be detected there. Their widths may be from centimeters to meters. They are usually composed of gouge, cataclasite, ultracataclasite or of their combination. Deformations can either be distributed uniformly over the fault core, or be localized inside a narrow shear zone. This zone of intensive grain grinding is the *principal slip zone* (PSZ). Its width is usually from one millimeter to decimeters [14, 16].

The structure of a fault zone depends on depth, properties of enclosing rock, tectonic conditions (shear, compression, tension), cumulated deformations, hydro-geological conditions and type of the deformation process. In slow aseismic creep the principal slip zone is often represented by a set of individual slip zones and zones of distributed shear deformations. Some secondary shears (they are often of oscillatory origin) can be localized along discrete fracture plains. The width of shear zones in the sections of aseismic creep of such faults as Hayward fault and San Andreas lie in the range of meters—tens of meters, the average value at the surface being 15 m

[14]. There are suggestions that this zone becomes narrower with depth and its width reduces to about 1 m deep in the rock massif.

An essentially higher degree of localization is observed in seismically active fault zones, where most deformations are, presumably, of coseismic origin. For example, investigations of shears in Punchbowl and San Gabriel faults in California have detected the thickness of the principal slip zone to be not wider than 1–10 cm. Note that cumulative displacements along these faults reach tens of kilometers [16, 71–73].

Chester and Chester [16] showed that it is in the principal slip zone that displacements of sides of large faults localize. According to their data, concerning one of the segments of the Punchbowl fault, only 100 m of displacement (of the total displacement length of 10 km) localized in the zone of fracturing about 100 m thick, while the rest occurred inside a narrow ultracataclasite layer from 4 cm to 1 m thick. A continuous, rather plane interface about 1 mm thick was detected inside this core. This interface was the principal slip zone of the last several kilometers of displacement [74].

In fault zones, whose cores consist of cataclastic rocks, coseismic ruptures often occur along one and the same interface, formed of ultracataclasites that have emerged at previous deformation stages [14]. Displacements along secondary, novel discontinuities are small and have a negligible contribution to the cumulative amplitude of fault side displacement.

Individual zones of the PSZ can rarely be traced longer than for several hundred meters, though it is suggested that their lengths can reach several kilometers [14]. In all likelihood PSZs may interact at some deformation stages through the zones of distributed cataclastic deformations without clear signs of a single rupture in latter. An analogy with laboratory sample destruction comes to mind here [75]. Such linear conglomerates of separate PSZs and sections of heterogeneous fracturing can make up an integrated fault core.

Geophysical investigations in wells that penetrate through fault zones at appreciable depths have also demonstrated an extreme degree of localization not only of deformation structure, but such parameters as porosity, permeability, velocity of propagation of elastic vibrations [19, 68].

Thus, the results of geological description of exhumed fault segments, the data on deep drilling of fault zones, as well as detailed investigations of seismic sources located with high accuracy [76] allow to speak about an extreme degree of localization of coseismic displacements. Macroscopic interblock displacements are not distributed over the thickness of material crushed in the course of shear, but are localized along a narrow interface of sliding. It means that with some conditionality, a dynamic movement along a fault can be considered as a relative displacement of two blocks, and their interaction is determined mainly by forces of friction.

4 Frictional Properties of Geomaterial and the Slip Mode

Investigating exhumed segments of fault zones has shown that the structural heterogeneity of large fault zones results in appreciable spatial variations in rheology and deformation rate at one and the same segment [77]. The evolution of frictional properties of rocks composing the massif and their spatial distribution play an important role in the processes of nucleation, localization and propagation of rupture in seismic events [78]. It has been shown in several works that the mineral composition affects the frictional strength and the sliding regime of the fault [79], and frictional stability depends on the evolution of structural properties of the fault during deformation [80].

Models that interpret emergence of different slip modes base on dependences of frictional properties of the sliding interface on velocity, displacement and P-T conditions revealed in laboratory experiments [81, 82].

These dependences are different for different geomaterials. Over the past years, a great number of experiments have been performed with materials collected while drilling fault zones. These experiments have shown that there exist materials with a pronounced property of velocity strengthening (VS). For example, saponite (a material with a low friction factor, which increases as sliding velocity grows), determines the deformation behavior of the creeping segment of the San Andreas fault [83]. Judging by the results of laboratory experiments weak materials rich with phyllosilicates manifest only stable sliding (corresponds to velocity strengthening), at least until the mineral composition of geomaterial in the principal slip zone alters with time. Stronger materials rich with quartz and feldspar, after creeping for a while, become 'velocity weakening' (VW) and provide unstable sliding [84].

VS- and VW-wise behavior can take place at different segments of one and the same fault zone. A complex topography of the fault interface leads to emergence of areas of stress concentration and rather unloaded areas. The probability of stick-slip to realize increases essentially in the areas of stress concentration. Deposition of minerals drawn by fluids takes place in unloaded areas, which in many cases promotes formation of layers composed of weak materials rich with phyllosilicates that manifest velocity strengthening. Thus, in many cases it is the contacts of rough surfaces (areas of stress concentration) that turn out to be dynamically unstable in sliding along a fault, while fault segments located between rough surfaces in contact manifest frictional properties of stable sliding. It is likely that these segments preserve their properties during, at least, several seismic cycles. This is supported by, the so called, repeated earthquakes (events identical in locations, energy and waveforms) [85]. These events repeatedly break again and again one and the same spot or asperity on the fault. These repeaters, whose manifestations are now found all around the world, have in most cases small magnitudes, but there are powerful ones too, with magnitudes higher than M6. These observations lead to the conclusion that different sliding regimes are determined by the non-uniform distribution of frictional properties and stress conditions over the fault interface.

It is convenient to demonstrate the effect of spatial non-uniformity of frictional properties on integral characteristics of the sliding process by the example of a

numerical calculation of the relative shear of two elastic blocks separated by a sliding interface. The friction between blocks was described by a rate-and-state friction law [81, 82]. According to relations of this empirical model, the coefficient of friction μ depends on the running velocity of sliding V and on the variable of state θ :

$$\mu = \mu_0 - a \ln\left(\frac{V_0}{V} + 1\right) + b \ln\left(\frac{V_0\theta}{D_c} + 1\right) \tag{4}$$

Here, μ_0 is the constant corresponding to stable sliding at a low velocity V_0 ; a, b, D_c are empirical constants, V is the running velocity of displacement, θ is the variable of state, which is determined by the kinetic equation:

$$\dot{\theta} = 1 - \left(\frac{|V|\theta}{D_c}\right) \tag{5}$$

When $(b - a) > 0$ the regime of velocity strengthening realizes. The case of $(b - a) < 0$ leads to velocity weakening and provides conditions for stick-slip to occur.

One or several spots with “true” condition of velocity weakening but different values of the friction parameter $\Delta = (b - a) < 0$ were “installed” into the sliding interface when the boundary conditions were set. A typical computation scheme is given in Fig. 3.

The rate-and-state law used here differed from the traditional one by the fact that after the instability has arisen the factors a and b in Eq. (4) were considered to be

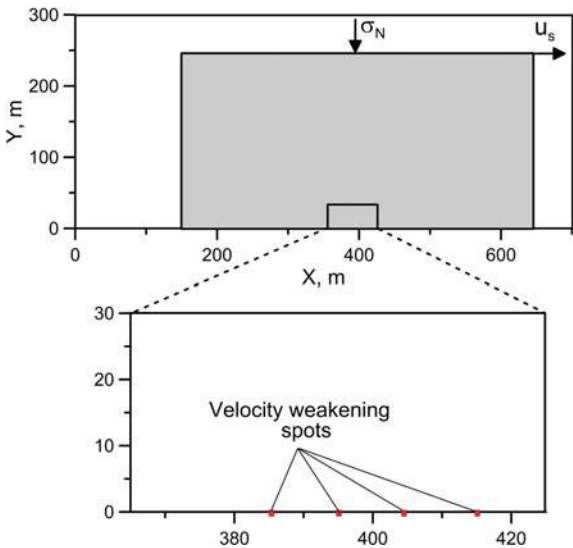


Fig. 3 Simulating the process of relative shear of two elastic blocks separated by a sliding interface: computation scheme. The block is pressed to the half-space by the normal stress. The upper right corner of the block is being pulled at a constant velocity u_s

zero. It was done to avoid repeated dynamic failures triggered by the waves reflected from the borders of the mesh.

At the rest sliding interface the force of friction was described either by the Coulomb's law with the same friction factor μ_0 (i.e., there was no dependence on velocity and displacement) or by law (4) with constants that provided velocity strengthening $\Delta = b - a > 0$. Kinematic parameters of motion, components of stress tensor, spatial distribution of alteration of power density of shear deformation of blocks, kinetic energy at different moments of time from the rupture start were controlled.

The size of the upper block (width, length) varied from 50×100 m to 200×600 m. The lower block was a half-space with the same elastic characteristics (the density 2.5 g/cm^3 , velocity of P-wave propagation $C_p = 3000 \text{ m/s}$, shear modulus $G = 52 \text{ MPa}$). The coefficients used in the main series of computations provided the regime of velocity weakening: $\mu_0 = 0.3$, $a = 0.0002$, $b = 0.0882$, $D_c = 1 \text{ } \mu\text{m}$, $V_0 = 0.002 \text{ mm/s}$.

Figure 4 shows the hodograph of a rupture propagating along a fault segment including 1 weakening spot. One can see that the rupture starts at a non-uniformity and propagates away from both sides of the spot. The velocity of rupture propagation along the interface of Coulomb friction is close to the one of P-wave, while at the interface with strengthening the velocity of rupture propagation is noticeably lower than that of the S-wave.

In the case of a contact with strengthening the amplitude of the velocity of displacement decays essentially faster in both directions. For example, at a distance of 100 diameters of the spot in the direction normal to the fault the maximum velocity of displacement in the case of interface with strengthening approximately 6 times lower than in the case of Coulomb friction.

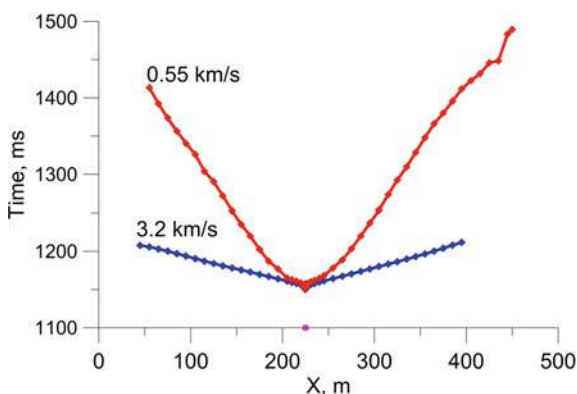


Fig. 4 Hodograph of the first onset of relative block motion. The size of the upper moveable block is 50×100 m. The lower block with the size of 400×800 m is immovable. Blue line— $\Delta = 0$ at the surface outside the spot; red line—velocity strengthening ($\Delta = 0.024$) is “switched on” at the surface outside the spot

Figure 5 shows the seismic moment $M_0 = G \cdot L \cdot U$ and kinetic energy of the block versus time for several variants of simulations. In the relation of seismic moment: L is the block length and U is the relative displacement. When friction between blocks increases as velocity and displacement grow (variant 2, 3), then the value of kinetic energy (the analogue of emitted energy) becomes lower than in the case of Coulomb friction. The final values of seismic moment are close for all the three variants, though the rate of growth of the value of M_0 is noticeably lower in the case of strengthening. It means that the energy of elastic deformation cumulated in the course of the interseismic period releases to a great extent during a rather slow sliding. It is this slow sliding that the dynamic rupture degenerates to, when it reaches the contact segment exhibiting velocity strengthening. This is well seen in the computation variant with several spots of weakening (Fig. 6). Again, the rupture starts at one of the spots of velocity weakening and propagates away from both sides of the spot. Outside the spot of weakening the velocity decays rapidly even in absence of segments with strengthening, speeding up again at neighboring VW spots. Though the maximum velocity of sliding decreases rapidly outside the spot of weakening, the integral value of relative ‘fault side’ displacement (the sum of dynamic slip and slow pre- and post-seismic slips) in this computation statement remains almost the same. The higher the total share of the VW spots is, the higher is the share of deformation energy spent to emission of the elastic wave in the high-frequency band of the spectrum.

For a great number of “spots-asperities” the rupture may start almost simultaneously at several spots. As far as the particle velocity is concerned, two types of segments can be clearly detected, just like in the case of several spots. The first one includes the spots themselves and their closest vicinity. Maximum velocity there reaches the values of 0.1–0.2 m/s. All over the rest contact the maximum particle velocity is 10–20 times lower.

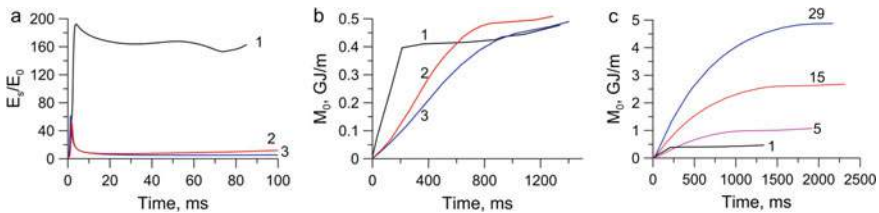


Fig. 5 Kinetic energy of the block and seismic moment versus time. The upper block is of the size of 150×450 m. **a, b** The centre of the single spot of velocity weakening of the size of $l = 1$ m is located in the point $x = 225$ m. 1—friction outside the spot doesn’t depend on velocity ($\Delta = 0$). 2, 3—friction outside the spot increases as the velocity and displacement grow ($\Delta = -0.008$ and $\Delta = -0.024$, relatively), **c** the number of velocity weakening spots is shown by corresponding numbers near the curves; parameters a and b in the R&S law vary in a random way in the limits: $a = 0.0002\text{--}0.0006$, $b = 0.0071\text{--}0.0084$

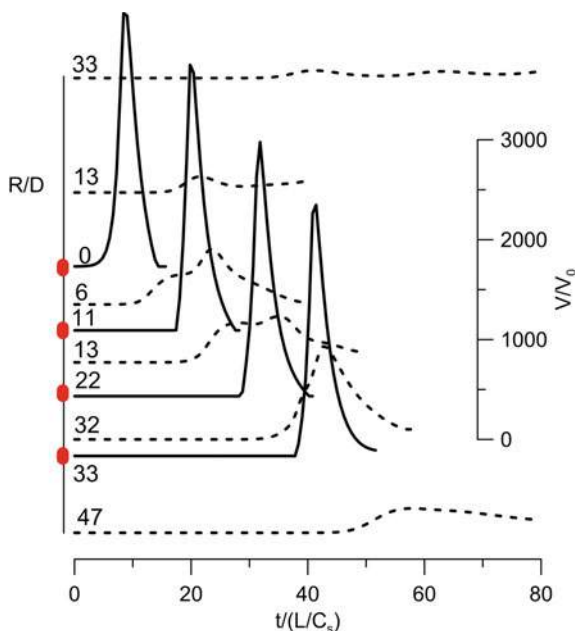


Fig. 6 Waveforms of particle velocity in the direction parallel to the sliding interface for the computation variant with four identical spots of velocity weakening. Locations of the VW spots are shown with bold segments at the left axis. Coulomb friction acts outside the spots. Numbers near the curves are the distances from the point of rupture start scaled by the spot diameter D . Solid lines are the epures corresponding to points inside the spots; dashed lines are the epures corresponding to points outside the spots. The amplitude of particle velocity is scaled by the velocity, at which the upper side of the block is pulled. Only first phases of motion are shown for better readability

Increasing quantity and density of asperities leads to an abrupt growth of the kinetic energy of block gained for the first 100 ms, i.e., the energy emitted in the high-frequency band of the spectrum. Meanwhile the integral value of seismic moment increases essentially slower.

Thus, the presence of VW spots determines the possibility of dynamic rupture to emerge. Their density and mutual locations govern the amount of energy emitted in the high-frequency band. Location and size of zones exhibiting velocity strengthening affect the velocity of rupture propagation, the scaled seismic energy (the ratio of E_s/M_0) and the size of the “earthquake”. Under the adopted statement of computations, termination of the dynamic rupture corresponds to a radical decrease of the velocity of sliding. The rupture propagates along a stressed tectonic fault to the zone, in which the contact exhibits velocity strengthening—velocity of displacement, emitted energy and rate of seismic moment growth harshly decrease. If the size of the strengthening zone is big enough, the “earthquake” degenerates into a “slow slip event” or the rupture terminates completely. The rupture crosses small zones, then again speeding up at VW spots. If there are no big strengthening segments at fault

interface, in this statement all the length of the block becomes involved in sliding, though in nature the rupture is usually terminated at some structural barrier [86].

5 Generating Different Slip Modes in Laboratory Experiments

Experiments on investigating regularities of emergence of different slip modes on a fracture with filler were performed on a well known slider-model set-up, in which a block under normal and shear loads slides along an interface (Fig. 7). A granite block (B) $8 \times 8 \times 3$ cm in size was put on an immovable granite base. The contact between rough surfaces (the average depth of roughness was 0.5–0.8 mm) was filled with a layer of discrete material (S), imitating the PSZ of a natural fault. The layer thickness was about 2.5 mm. The normal load σ_n was applied to the block through a thrust bearing. σ_n varied in the range of 1.2×10^4 to 1.5×10^5 Pa. The shear load τ_s was applied to the block through a spring block. Its stiffness could vary. The set-up was equipped with an electromotor with a reducer that allowed to maintain the velocity of loading u_s with high accuracy in the range of 0.08–25 $\mu\text{m/s}$. The shear force was controlled with a force sensor. Displacements of the block relative the base were measured with LVDT sensors with the accuracy of 1 μm and with laser sensors (D) in the frequency range of 0–5 kHz and with the accuracy of 0.1 μm .

Filling the contact with mixtures of different materials, we managed to realize a wide spectrum of slip modes in the experiments, which correspond qualitatively to all types of interblock movements observed in nature—from aseismic creep to earthquakes. Examples of slip episodes realized in experiments are shown in Fig. 8. The figure also shows the recorded pulses of acoustic emission radiated during sliding.

Let's consider the conditions for slip to occur. A necessary condition for a slip to occur is closeness of effective stresses, tangential to fault plane, to the local or running ultimate strength:

$$\tau \geq \tau_0 \quad (6)$$

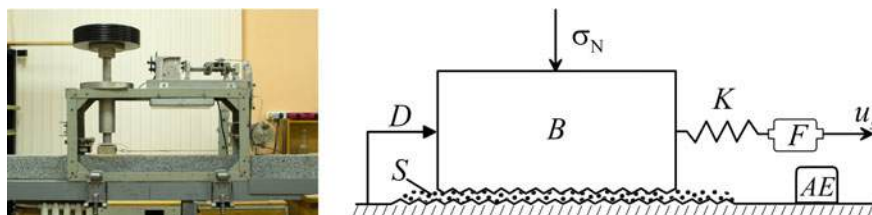


Fig. 7 Photo of the slider-model setup and the scheme of the experiments. B —movable block, S —layer of filler, D —laser sensor, K —spring element, F —force sensor, AE —acoustic emission sensor

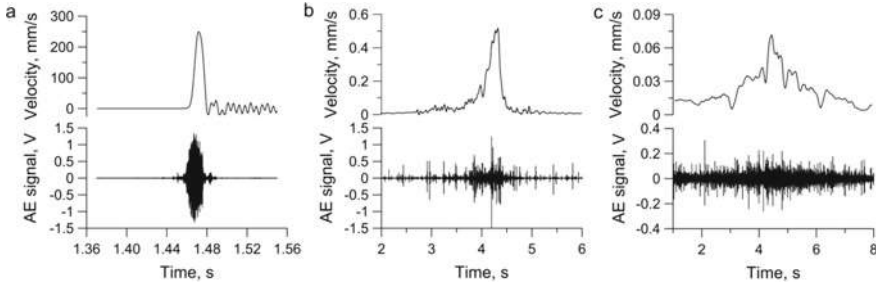


Fig. 8 Examples of diagrams of block velocity corresponding to different deformation events and their acoustic signal portraits. **a** fast mode (corresponds to normal earthquake); **b** medium mode (corresponds to slow earthquake); **c** slow mode (corresponds to slow slip event)

We use the term “local ultimate strength” because slip can occur under tangential stresses knowingly lower than the Coulomb strength τ_p .

Another necessary condition is the weakening of the sliding area as the velocity v and/or amplitude D of fault side displacement grows:

$$\frac{\partial \tau}{\partial v} < 0; \quad \frac{\partial \tau}{\partial D} < 0 \quad (7)$$

It is clear that if the contact strength will not decrease during shear, a dynamic slip will be actually impossible.

And, at last, the third condition: the rate of decrease of stresses in the enclosing massif, tangential to sliding interface (this is the shear stiffness of the massif K) should be lower than the rate of decrease of resistance to shear (this is the modulus of fault stiffness k_s at the post-critical section of the rheologic curve):

$$|k_s| = \left| \frac{\partial \tau}{\partial D} \right| \geq K = \eta \frac{G}{\hat{L}} \quad (8)$$

G is the shear modulus of the enclosing massif, $\eta \sim 1$ is the shape factor [75], and \hat{L} is a specific size linked to the magnitude of the earthquake.

It is convenient to demonstrate the meaning of relations (6)–(8) at the scheme shown in Fig. 9. After the stress has reached the running strength of the contact (τ_p corresponds to the maximum of rheological curve $\tau(D)$ in Fig. 9, though it is not obligatory) the contact strength starts to decrease as the relative displacement and velocity grow. If condition (8) is true (black solid line in Fig. 9), a dynamic instability occurs and the energy is emitted outside the system. The amount of emitted energy in this simple example corresponds to the area bounded by the rheologic curve and the solid line of unloading of the enclosing massif. In the case when condition (8) is not true (dashed line in Fig. 9) the dynamic slip and, consequently, emission of energy are impossible.

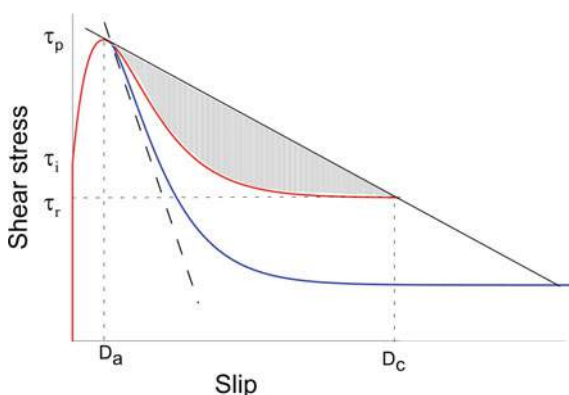


Fig. 9 Scheme of the slip emerging on a fault. Red line is the rheologic dependence stress-displacement; black straight lines are the lines of enclosing massif unloading in the course of relative fault side displacement. Black solid line—the rate of massif unloading is lower than the rate of decrease of resistance to shear along the fault. This case corresponds to emergence of dynamic slip. The energy emitted by a unit area of the sliding interface corresponds to the hatched area. Black dashed line—the rate of massif unloading is higher than the rate of decrease of resistance to shear along the fault. The fault remains stable in this case. Blue line is the decrease of effective friction during sliding at the expense of frictional melting, thermal effects, etc. It is well seen that these processes can change the parameters of an earthquake, but not the moment of dynamic slip start

In the course of sliding at velocities of $\sim 1\text{--}10$ m/s and in P-T conditions specific for seismogenic depths, a number of processes severely affecting the parameters of resistance to shear can occur at the interacting surfaces. These are the effects of friction lowering either because of thermal effects or due to effects produced by the high velocity of sliding: frictional melting [87], dynamic lubrication with solid materials [88], localization of heating in the area of “real” contact [89], macroscopic rise of temperature and the effect of velocity weakening [90], thermal decomposition of minerals leading to growth of pore pressure and generation of weak material [91], generation of silica gel during quartz amorphization under high pressure and large deformations [92], neo-mineralization of the sliding interface at the nano-crystal level [93].

All these effects are very important because they lead to a decrease of residual friction τ_r (blue line in Fig. 9) and, consequently, a decrease of the amplitude of stress drop $\Delta\tau$ and of the amount of emitted energy (magnitude of the earthquake). However, as these phenomena have no effect on fulfillment of conditions (6)–(8), they are of no use in searching for the signs of dynamic failure preparation.

Thus, it seems reasonable to select the modulus of the rate of decrease of resistance to shear (fault stiffness $|k_s|$ at the beginning of the post-critical section of the rheologic curve) as the characteristics which controls the initial stage of earthquake rupture nucleation. The ratio $\psi = |k_s|/K$ determines not only the possibility of sliding, but its character as well.

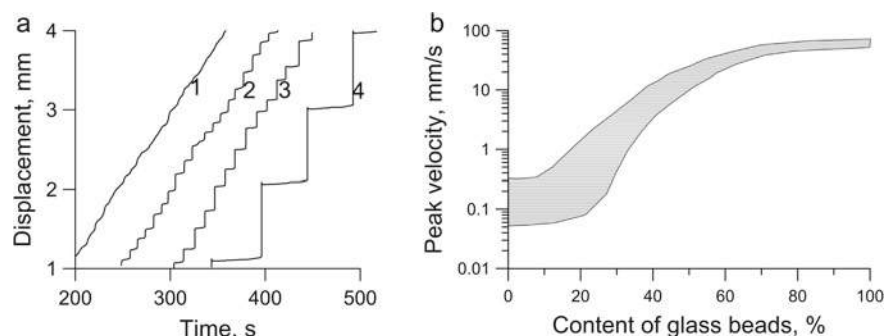


Fig. 10 Transformation of sliding regime. **a** Displacement versus time for experiments with filler consisting of quartz sand with different admixtures of glass beads. (1) pure sand, (2) 20% of glass beads, (3) 40% of glass beads, (4) pure glass beads. **b** The variations in peak velocity versus changes in the gouge texture

A certain slip mode realized in fault deformation is determined by structural, physical and mechanical properties of the filler. In a regular stick-slip repeated slip events take place with close parameters, while in irregular regimes stochastic events are observed, and their statistics obeys a power law.

For example, increasing the share of smooth grains (glass beads) leads to jamming of the granular layer and transition from stable sliding to stick-slip (Fig. 10). Earlier a similar result was obtained in [94]. In absence of smooth grains 90% of all the events have peak velocity of only 0.1–0.2 mm/s, while events with peak velocities exceeding 0.5 mm/s are absolutely absent. Thus, we can say that the steady sliding mode is realized.

When the share of glass beads reaches 40–50% of filler mass the motion becomes the stick slip with a relatively small value of stress drop. A further increase of the amount of smooth grains changes only the amplitude of displacement during a failure. When the filler consists only of glass beads the value of stress drop $\Delta\tau$ reaches approximately 15% of the maximum value of shear strength τ_0 , and the peak velocity of blocks reaches the value of 60–80 mm/s.

The emergence of a certain slip mode is determined not only by the grain geometry, but also by their chemical and physical properties. For example, for mono-component low dispersion fillers consisting of angulated grains with ionic bonds between molecules (sodium chloride, corundum, magnesium oxide) realization of dynamic failures is much more probable than for fillers with covalent or metallic bonds between molecules (dry quartz sand, graphite and others) under similar loading parameters [95]. Probably the ionic bonds provide a stronger adhesive interaction of the filler grains in contact.

Moistening of the filler has an essential effect on the slip mode. It is well known that adding even a small amount of liquid to a granular media changes its collective stability [96, 97]. It is interesting that the value of stress drop changes non-monotonically as the viscosity of the moistening fluid grows (Fig. 11).

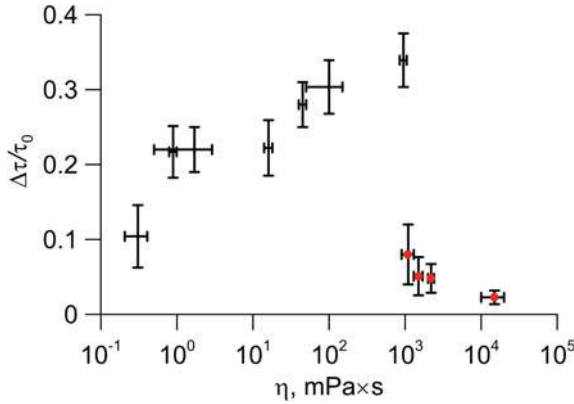


Fig. 11 Variations of shear stress drop versus fluid viscosity. Red symbols correspond to experiments with post-critical values of viscosity

Varying interstitial fluid viscosity, we traced the transformation of the sliding regime. The mass humidity of gouge was 1%. Variation of liquid viscosity η from 3×10^{-4} to 2×10^1 Pa s caused a 15 times change of static stress drop. At the same time relative change of τ_s was only 25%. The value of $\Delta\tau/\tau_s$ gradually increases with increasing liquid viscosity up to the critical value $\eta_{cr} \approx 1$ Pa s and then sharply drops by almost one order of magnitude, after which the value of $\Delta\tau/\tau_s$ gradually decreases with increasing viscosity up to 20 Pa s. At the range of after-critical values of viscosity the peak velocity decrease is approximately inversely proportional to the value of viscosity, and growth of slip duration is observed.

The presence of a small amount of interstitial liquid promotes jamming of the model fault. Probably, fault jamming is caused by the emergence of an additional intergrain force. The higher the force is, the higher is the value of elastic energy accumulated and, consequently, the higher are the stress drop and peak velocity during the slip episode. The cohesive force between gouge grains initially increases sharply for very small volumes of liquid because of the roughness of grains, further cohesiveness is effectively constant at least up to the stage when the liquid occupies about 35 percent of the available pore space [98, 99].

During the slip episode a reorganization of the meso-scale fault structure takes place, manifested in intergrain slipping. [95, 100]. At the stage of rest, when velocity is low, the intergrain contact dewets, which is accompanied by the accumulation of excess liquid in the pore space. But during the slip episode the liquid can penetrate into the grain contact at “critical” slip velocity conditions [101]. Probably, in the presented experiments, in the case of pre-critical viscosity ($\eta \leq 1$ Pa s) the slip velocity is not high enough for the “lubrication effect” to occur. But when $\eta > 1$ Pa s liquid penetration takes place, provoking damping of stick-slip events and the higher value of viscosity corresponds to the lower value of peak velocity. According to [102], when the fluid viscosity is about 10^5 Pa s, the occurrence of steady sliding must be observed.

The effect of filler properties on the sliding regime can be traced if one uses low dispersion mixtures of different grain materials as fracture filler. Variations of clay content in the moistening sand/clay mixture provided considerable changes of the fault slip modes. Figure 12 shows the dependences of slip episode parameters for different content of clay in the gouge moistened with water and glycerol. As clay content was increasing, transformation from stick-slip to steady sliding was observed. For the clay content of less than 15% a regular regime of fast slip episodes took place. Increasing clay content up to 20% led to the formation of irregular regime consisting of fast and slow slip episodes. The average values of slip episode parameters varied relatively slow (about an order of magnitude) in the range of clay content of $v = 0$ –25%. When clay content was increased from 25 to 28% of mass, the decrease of average value of peak velocity by more than two orders of magnitude (from ~ 7 mm/s to ~ 0.04 mm/s) occurred and shear stress drop decreased 5 times. For mixtures with clay content of 30% rare slow slip episodes were observed, and for mixtures with clay content exceeding 35% the sliding became steady.

A similar pattern of changes in the fault behavior, but more drastic, was observed in experiments with glycerol. In the range of clay content of $v = 0$ –1% the peak velocity decreased 3 times, further increase of clay content from 1 to 3% resulted in the drop of peak velocity by approximately 3 orders of magnitude. For mixtures with clay content exceeding 5% sliding became steady.

A wide spectrum of shear deformation regimes was realized in the presented experiments by changing the filler properties or the stiffness of loading—from dynamic failures to stable sliding. It should be emphasized that slow slip modes have all the phases intrinsic for stick-slip—acceleration, prolonged sliding, braking, stopping and the phase of rest. This suggests the idea that all the slip modes on faults span a continuum—they may be considered as a single set of phenomena.

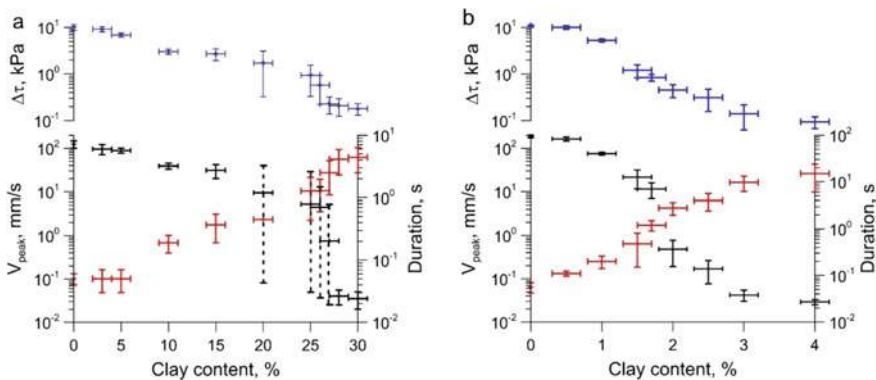


Fig. 12 Variation of shear stress drop (blue), peak slip velocity (black) and the slip episode duration (red) versus clay content in the gouge moistened with water (a) and glycerol (b). Vertical solid lines are standard deviations, vertical dashed lines show the range of variation of slip event parameters for irregular slip regimes

Interesting consequences follow from comparison of the laboratory experimental data to results of numerical simulation basing on the rate-and-state frictional law—Eqs. (4) and (5). By overrunning the constants in the rate-and-state relations a , b and D_c the best fit of simulated and experimental data was provided.

Figure 13a compares the simulated and experimental dependences of block slip velocity on time for the contact filled with quartz sand. For convenience, here and below the time is counted from the moment of velocity maximum. The following parameters of the rate-and-state model were used in simulations: $a = 0.0002$, $b = 0.00109$, $D_c = 10 \text{ }\mu\text{m}$. The characteristics of slider model corresponded to experiment. The figure confirms that in the case of a pronounced stick-slip the simulated epure reproduces the experimental one rather well. Comparing the results of simulations to the experiment at the diagram force-displacement also demonstrates good correspondence. Calculations made for numerous experiments have shown that failure episodes during stick-slip (“laboratory earthquakes”) can be satisfactorily simulated with the canonical rate-and-state law (4).

Attempts to simulate the slow sliding regimes (laboratory “slow earthquakes” and “episodes of slow slip events”) have met certain difficulties. This is well seen in the example presented in Fig. 13b, which shows the results of calculation of the block slipping slowly along the contact filled with dry clay, the stiffness of loading element being equal to 17 kN/m. In this experiment the duration of sliding has increased, if compared to the contact filled with dry quartz sand, approximately by an order of the magnitude, along with a corresponding decrease of both the maximal and the average slip velocities (curve 1 in Fig. 13b).

We failed to reproduce such a signal in simulations involving only the canonical rate-and-state model (curve 2 in Fig. 13b). Varying the parameters, one can only fit the amplitude of the signal, but the “fullness” of the epure, i.e., the value of displacement, is controlled mainly by block mass and spring stiffness (parameters

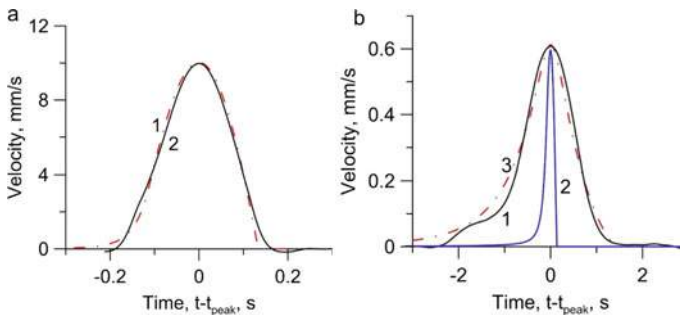


Fig. 13 **a** Block velocity versus time for the contact filled with quartz sand. R&S parameters of simulation: $K = 17 \text{ kN/m}$; $\mu = 0.61$; $D_c = 10 \text{ }\mu\text{m}$; $a = 2 \times 10^{-4}$; $b = 1.09 \times 10^{-3}$. Solid line is the simulation, dashed line is the experiment. **b** Block velocity versus time. The contact is filled with dry fire clay. The stiffness of the spring is $K = 17 \text{ kN/m}$; (1)—experiment; (2)—simulation with $\eta_d = 0$ in Eq. (9); (3)—simulation with $\eta_d = 1000 \text{ Pa s}$. Parameters of the R&S model: $\Delta_c = 6.5 \times 10^{-4}$; $\mu = 0.7$; $D_c = 50 \text{ }\mu\text{m}$; $a = 1.2 \times 10^{-4}$; $b = 1.35 \times 10^{-3}$

rigorously defined for each experiment). We could only “dilute” the onset of the peak by changing D_c . The simulated peak of the dynamic failure for a given amplitude always remained essentially narrower than the one obtained in experiment.

We managed to better fit the results of simulating “slow” movements to experiment by introducing a term to the canonical R&S Eq. (4) that takes into account the emergence of additional resistance to shear produced by the “dynamic viscosity” of the contact:

$$F_s = \sigma_N \cdot S \cdot \left[\mu_0 + a \ln\left(\frac{|\dot{x}|}{u^*}\right) + b \ln\left(\frac{u^* \theta}{D_c}\right) \right] + \frac{\eta_d \cdot S \cdot \dot{x}}{d} \quad (9)$$

where η_d is the factor of dynamic viscosity of interblock contact; S and d are the area and thickness of the contact zone. Results of simulations with account for the dynamic viscosity are presented in Fig. 13b by the line 3. One can see that introducing the “dynamic viscosity” of the contact into the equation of motion allows to reproduce the slow slip mode in simulations with satisfactory accuracy, too.

Equation (9) allowed to satisfactorily reproduce the character of motion in all the experiments by fitting the effective viscosity. Figure 14 gives examples of eures of slip velocities of the block in two experiments—with the contact filled with watered fire clay and with the contact filled with quartz sand with an admixture of 25% of talc. In both cases the slippages have long phases (~30–40 s) of gradual velocity increase, and then phases of deceleration of about the same duration. The maximum velocity of displacement decreases to 10–100 $\mu\text{m/s}$. For best fitting of simulated and experimental results the factor of effective viscosity η was increased by more than an order of a magnitude if compared to the simulation of experiment with dry fire clay (Fig. 13b).

It is well known that the generalized viscosity of a rigid body is not the property of the material itself, as for example, the viscosity of Newtonian fluid. This parameter

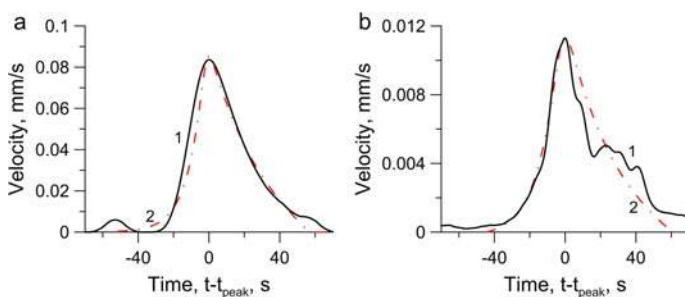


Fig. 14 Block velocity versus time for the slow slip mode. **a** contact filled with watered fire clay; (1) experiment; (2) simulation; Parameters of modified R&S model: $K = 10 \text{ kN/m}$; $\Delta_c = 6.5 \times 10^{-4}$; $\mu = 0.56$; $D_c = 90 \text{ MKM}$; $a = 1.0 \times 10^{-4}$; $b = 3.5 \times 10^{-3}$; $\eta_d = 2.76 \times 10^4 \text{ Pa s}$. **b** contact filled with mixture of quartz sand (75%) and talc (25%). (1) experiment; (2) simulation; Parameters of modified R&S model (9): $K = 16.56 \text{ kN/m}$; $\Delta_c = 2.3 \cdot 10^{-4}$; $\mu = 0.61$; $D_c = 18 \text{ }\mu\text{m}$; $a = 1.0 \times 10^{-5}$; $b = 8.5 \times 10^{-4}$; $\eta_d = 3.94 \times 10^4 \text{ Pa s}$

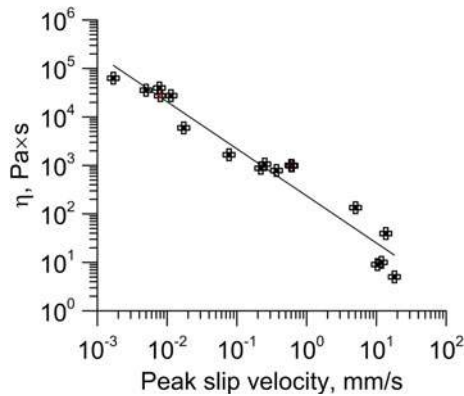


Fig. 15 Effective viscosity versus peak velocity of block displacement. The line shows the best fit by the Eq. (10) with $R^2 = 0.94$

is a characteristics of the rigid body rheology, depending on the specific time of deformation process, or, to be more precise, on the deformation rate.

Figure 15 shows the dependence of the factor of effective viscosity η_d on the peak velocity of block displacement. The effective viscosity was obtained by fitting the results of simulations to the ones of experiments, the simulations being performed according to the modified Rate-and-State law (9). Symbols show the results for different fillers and stiffnesses of the loading device. The line shows the best fit by the following power function:

$$\eta_d = 235 \cdot u_{\max}^{-0.97} \quad (10)$$

Thus, the conducted laboratory and numerical experiments have demonstrated that the effective viscosity is a conventional parameter with the dimensionality of Pa s. This parameter is convenient to describe the alterations of fault slip modes. The factor of dynamic viscosity depends both on the properties of the filler and on the stiffness of the loading system. A verification of the results of simulations involving the rate-and-state model allows to conclude that to have the opportunity of modeling all the spectrum of fault slip modes the empirical Dieterich's law [81, 82] should be supplemented by a term that takes into account the emergence of additional resistance to shear produced by the dynamic viscosity of the fault interface, or, to be more precise, by the fact that the force of resistance to shear depends on the velocity of interblock motion. After this the episodes of slow slip observed in experiment can be reproduced with an appropriate accuracy. The obtained results agree with the data presented in the works [103, 104], where the dependence of viscosity factor on specific time of deformation is close to a linear one.

6 Radiation Efficiency of Slip Episodes

It is evident that in different slip modes different shares of deformation energy are emitted as seismic waves. It is a common thing that seismic events produced by the processes of deformation and destruction of a rock massif are described by two parameters: seismic moment (M_0) and seismic energy (E_s). The scalar seismic moment:

$$M_0 = \mu \cdot D_s \cdot S \quad (11)$$

is a generally recognized measure of the event size. This value does not depend on details of the process in the source, because it is determined by the asymptotics of the spectrum of displacements in the low-frequency band. It is proportional to the amplitude of the low-frequency band of the spectrum, and, provided that up-to-date equipment and processing methods are used, it can be estimated rather reliably. In Eq. (11), μ is the shear modulus of rock in the source, D_s is the displacement along the rupture, S is the area of the source. The divergence of estimations made by different authors for powerful earthquakes rarely exceeds 2–3 times.

The seismic energy E_s , i.e., the part of the deformation energy emitted in the form of seismic waves, on the contrary, is determined by the dynamics of rupture development and depends on velocity of rupture propagation, balance of energy in the source, etc. The value of seismic energy is usually determined by integrating the recorded vibrations.

In laboratory experiments the following product can be considered as the analogue of seismic moment: $M_{lab} = K \cdot D \cdot L$. Indeed, in nature the seismic moment is the product of shear force drop ΔF_s acting at the fault plane by the fault length L_s :

$$M_0 = \mu \cdot D_s \cdot L_s^2 = \mu \frac{D}{L_s} L_s^3 = \Delta \tau \cdot S \cdot L_s = \Delta F_s \cdot L_s \quad (12)$$

In this relation D_s is the displacement produced by slip events, $S \approx L_s^2$ is the area of rupture, $\Delta \tau$ is the tangential stress drop at the fault due to slip. For laboratory experiments with the spring-block model the single shear force drop can be written as $\Delta F = K \cdot D$, where K is the spring stiffness, D is the block displacement.

Then, comparing the energy budgets for earthquake and for laboratory slip event [23], we may consider the relation $e_{lab} = E_k / (K \cdot D \cdot l)$ as the analogue of the energy/moment ratio $e = E_s / M_0$, which is used in seismology to characterize the seismic efficiency of an earthquake. Here l is the block size.

It is convenient to estimate the share of deformation energy that transited to the kinetic energy of block motion (emitted in a laboratory “earthquake”) using the experimental dependence of the measured shear force on block displacement. Examples of such dependences are given in Fig. 16.

When stresses, tangential to fault interface, reach the ultimate strength of the contact τ_0 and the condition (6) is true, the resistance to shear $\tau_{fr}(x) = R(x)/l^2$

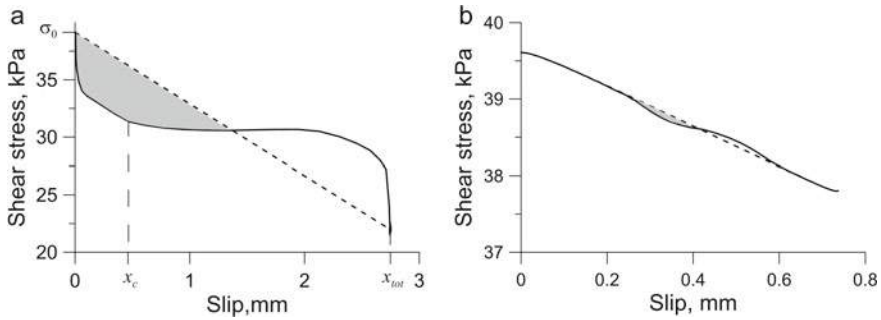


Fig. 16 Examples of experimental diagrams ‘shear stress—displacement’. Solid line is the frictional resistance of the contact. Dashed line is the force applied by the spring. Shaded areas correspond to the values of kinetic energy of the block. **a** Fast mode (laboratory earthquake). The filler is quartz sand moistened with glycerol (1% of the mass). **b** Slow mode (laboratory slow slip event). The filler is watered mixture of quartz sand (70%) and clay (30%)

starts to decrease with displacement faster than the applied load $\tau_s(x) = F_s(x)/l^2$. As a result a slip starts, which is described by the equation:

$$m \cdot \frac{\partial^2 x}{\partial t^2} = [\tau_s(x) - \tau_{fr}(x)] \cdot l^2 \quad (13)$$

where m is the mass of moveable block, x is the relative displacement of blocks. After a certain displacement D_c has been reached (its value depends on roughness of fault walls, filler properties, etc.) the value of τ_{fr} comes to the residual (dynamic) value and stops changing.

After the following condition becomes true:

$$\int_0^x [\sigma_s(\chi) - \tau_{fr}(\chi)] d\chi = 0 \quad (14)$$

block slippage along the interface ceases ($x_{tot} = D_{tot}$ in Fig. 16a) and a new cycle of cumulating the deformation energy starts.

If the character of the dependence $\tau_{fr}(x)$ is such that the difference $\tau_s(x) - \tau_{fr}(x)$ becomes negative earlier than the minimum possible value of the friction force is reached (Fig. 16b), the contact actually doesn’t come to the slip regime fully, which results in a small stress drop. Such an effect is most pronounced for fractures with high content of ductile grains (clay, talc).

The energy E_s , emitted during a slip event, was determined by integrating the difference of experimental dependences of the applied load $\tau_s(x)$ and resistance to shear $\tau_{fr}(x)$:

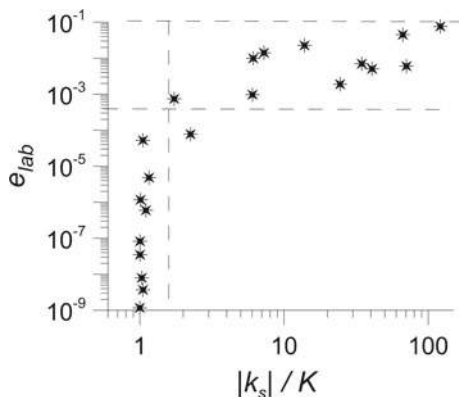


Fig. 17 Scaled emitted energy versus ratio of the stiffnesses of fault to enclosing massif. Symbols are the results of laboratory experiments. The area bounded by two horizontal dashed lines corresponds to dynamic failures. Slow events are to the left from the vertical dashed line

$$E_s = \int_0^{D_0} [\tau_s(\zeta) - \tau_{fr}(\zeta)] \cdot l^2 \cdot d\zeta \quad (15)$$

where D_0 is the displacement, for which $\tau(D_0) - \tau_{fr}(D_0) = 0$. Shaded areas in the examples presented in Fig. 16 correspond to the values of E_s .

Thus, the ratio of the stiffness of fault to the one of enclosing massif (fracture and spring in laboratory experiment) $\psi = |k_s|/K$ determines not only the possibility of slippage, but its character as well. The dependence of scaled emitted energy e_{lab} on this parameter, plotted according to the results of laboratory experiments is shown in Fig. 17.

It is well seen that stick-slip takes place in a rather wide range of the values of ψ , while slow slip modes realize in a narrow area of the values of $|k_s|/K \sim 1 \div 2$. It means that at “brittle” faults, whose stiffnesses (the rate of decrease of resistance to shear) are rather high, the deformation energy releases solely in dynamic failures—normal earthquakes. Slow slip events can take place at faults with low stiffnesses. As massif stiffness in the crust alters slightly for different regions and different depths, it is the fault stiffness k_s that should be taken as governing parameter.

7 On Artificial Transformation of the Slip Mode

The main anthropogenic factors that can trigger movements on a prepared fault are variations of fluid-dynamic regime in the fault zone, effect of seismic vibrations, excavation and displacement of large amounts of rock in mining. Irrespective of what factor we are speaking about, the geomechanical criteria (6)–(8) formulated

above should be true at a certain fault segment and in the enclosing rock massif. It makes sense to assume that the specific size of the segment should exceed the size of the, so called, “zone of earthquake nucleation”—the section where the rupture rate reaches the dynamic value [75]. Currently this value can be estimated only roughly. According to seismological data [103–107], the size of nucleation zone L_n can reach about 10% of the length of future rupture, i.e., for $M = 6$ $L_n \leq 1000$ m. Let us consider possibilities of realization of the formulated criteria under anthropogenic factors.

7.1 *Changing the Fluid Dynamics*

The effect of injection/withdrawal of fluid in/out of rock masses on seismicity has been studied in numerous works. One can find citations of corresponding publications, for example, in monographs and reviews [108, 109]. Rising pore or formation pressure and corresponding decrease of the effective Coulomb strength of faults and fractures is considered to be the main physical mechanism. However, there are many evidences [109, 110] that very weak variations of hydrostatic pressure (about millibar) affect seismicity. It is unlikely that the Coulomb model can be applicable here. It has been established in in situ experiments that the size of the area where parameters of sliding regime along a fault change can exceed several times the radius of the zone of pore pressure alterations [57]. It means that injection or withdrawal of fluid can change the characteristics of geomaterial.

The frictional parameter—the difference $(a - b)$ from Eq. (4)—decreases abruptly, i.e., velocity weakening becomes more pronounced even when a small portion of fluid is injected. In the laboratory experiments described above adding fluid weighing 0.1% of the mass of laboratory fault filler is enough for a radical change of the character of sliding from creep to pronounced stick-slip [65].

Injecting fluid is, probably, one of the few possibilities to change the frictional parameter in situ. Such an effect was observed in the above laboratory experiments, in which the stick-slip of a granite block on a thin layer of granular material was investigated. Increasing fluid content, when its volume share had already reached $\zeta \approx 0.1\%$, resulted in a rather abrupt transition from stable sliding to stick-slip. In the presence of glycerol the maximum velocity of sliding increased more than 300 times. In case the humidity was further increased the regime stabilized and up to $\zeta \approx 10\%$ the deformation regime exhibited almost no dependence on content of fluid in the filler. Most likely, this phenomenon results from the character of interaction of particles in the fracture filler. After adding a small amount of fluid a thin film of sub-micron thickness emerged at the surfaces of filler grains. This film smoothed roughness and promoted good contacts between separate grains. The deformation regime on the fracture depends essentially on the effective viscosity of the fluid (Fig. 11). The performed estimations show that in nature colloidal films covering filler grains may emerge during the processes of aggregation—formation of enlarged structural elements as a result of adhesion of separate grains. Judging by the results

of the performed experiments, viscosity of these films, i.e., the chemical content of clays, may affect the regime of fault deformation.

In laboratory tests described in [111], fast injection of fluid into the contact zone resulted in alteration of such sliding parameters as velocity of movement, stress drop, emitted energy. Alteration of pore pressure was negligible in comparison to the normal stress on the fault, i.e., it is the changed frictional properties of the contact that led to alteration of the slip mode. It should be emphasized that in this case the effect was observed after an appreciable amount of fluid had been injected ($\sim 20\%$ of fracture volume, the porosity of the filler being about 35%), so, the fluid spread to about 80% of the contact area.

Injecting noticeably less amount of water in experiments on loading a monolithic heterogeneous sample [112] resulted in appreciable variations of the regime of acoustic emission and kinetics of the process of macro-destruction. These processes, probably, caused by physical and chemical interactions in fracture snouts of the type of Rehbinder effect, had no relation to the effect of alteration of parameters of velocity weakening during sliding, being discussed here.

Thus, though an anthropogenic change of fluid dynamics hypothetically can lead to triggering a dynamic movement, one should keep in mind that this change should involve a rather big fault area.

7.2 *Effect of Seismic Vibrations*

Triggering seismic events by vibrations of earthquakes that occurred at distances of hundreds and thousands of kilometers is an admitted example of the trigger effect [18]. As far as the data of numerous investigations of the so called “dynamic triggering” is concerned, we should note that in most cases the minimum strain level needed for initiation is estimated to be about $\sim 5 \times 10^{-7}$ – 10^{-6} , though some authors give less estimations [113]. In most cases occurrence of dynamically triggered seismicity is linked to the effect of low-frequency surface waves with periods of 20–40 s. It is generally admitted that triggering with high-frequency body waves seems hardly probable. A detailed review of this topic is given, for example, in [18] and in the monograph [65].

Explosions turn out to be less effective from the point of view of triggering dynamic movements, than powerful distant earthquakes. The results of measuring parameters of seismic vibrations produced by ripple-fired explosions (for example, in [112]) allow to estimate the maximum particle velocity of the wave at different distances. For delay-fired explosions with typical parameters used in mining, maximum particle velocity at the distance of $R \sim 3\div 5$ km does not exceed the value of $V_m \sim 0.3\text{--}0.6$ mm/s. The characteristic frequency of vibrations is about $f \sim 0.1\text{--}0.5$ Hz, while the duration of the wave-train may reach 100 s. It should be emphasized that increasing the integral energy of a delay-fired explosion leads (at distances of several kilometers) only to an increase of duration of the signal, but actually has no effect on the values of peak ground velocity (PGV). Therefore, dynamic stresses in

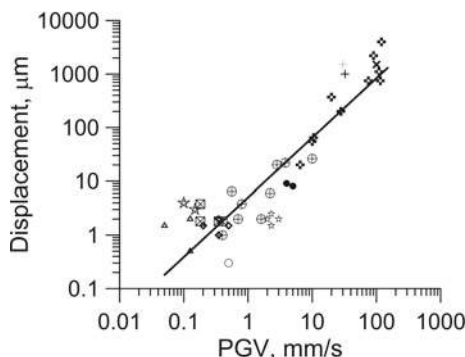


Fig. 18 Residual displacements registered on discontinuities versus the value of peak ground velocity (PGV) in seismic wave

seismic waves at the depth of 3–5 km can reach the values of only several kPa, and strains— $\sim 10^{-7}$.

Estimations [65] and the data of precise measurements of residual displacements on faults, produced by seismic vibrations from explosions, show (Fig. 18) that the expected value of a residual displacement on a fault under the effect of such a disturbance can be from sub-microns to tens of microns, and only in extreme cases it can reach 1 mm. Under such displacements a direct triggering of earthquake of an appreciable magnitude by seismic waves from explosions is hardly to be expected, because according to seismological data the value of critical displacement for an earthquake of average size ($M \sim 6$) is about 10 cm [75, 105]. It can be seen in Fig. 18 that for a noticeable effect to occur the PGV at an appreciable part of the fault area should be about 10 cm/s and even more. It means that the charge should be installed rather close to a fault that have reached the ultimately stressed state. An example of such an event is the April 16, 1989 earthquake in Khibiny Mountains [113].

7.3 *Excavation and Displacement of Rock in Mining*

Perhaps the most powerful anthropogenic triggering factor is the displacement of rock in mining. Considering this question is out of the frames of this article. We should only note that excavation of material from a large operating quarry with the sizes of kilometers in plane and hundreds of meters in depth leads to a reduction of Coulomb stresses up to 1 MPa at the planes of faults located at depths of several kilometers [114]. This value is negligible compared to the level of lithostatic stresses. But it may turn to be enough to trigger a seismogenic movement along a stressed fault. This is supported by the well known calculations of the field of static stresses in the vicinity of hypocenters of aftershocks of powerful earthquakes [115]. It is important to emphasize that for large quarries the size of the zone, in which the

change of Coulomb stresses at fault plane exceeds 10^{-1} MPa, is essentially bigger than the size of the nucleation zone of an earthquake with magnitude $M \leq 6$.

Open (surface) mining operations in most cases only bring the moment of the earthquake closer, while underground developing of deposits changes effective elastic moduli of the rock massif in the vicinity of an active fault [116]. Therefore, it seems probable that without the anthropogenic effect the cumulated deformation energy would have released not through a dynamic movement (earthquake), but in another way, e.g., through slow creep or a series of slow slip events.

8 Conclusion

Discovery and classification of sliding regimes on faults and fractures, that are transitional from stable sliding (creep) to dynamic failure (earthquake), alter to a great extent the understanding of how the energy cumulated in the process of the Earth's crust deformation is released. Slow movements along faults are now perceived not as a special sort of deformations, but together with earthquakes span a continuum of slip modes.

Judging by the results of laboratory tests, small variations of material content of the fault principal slip zone can lead to an appreciable change of the part of seismic energy emitted in dynamic unloading of the adjacent section of the rock massif. Regimes of interblock sliding with values of scaled kinetic energy differing by several orders of magnitude, while differences of contact strengths are small, have been reproduced in experiments. The obtained results allow to conclude that the sliding regime and, in particular, the part of deformation energy that goes to seismic emission, is determined by the ratio of two parameters – stiffness of the fault and stiffness of the enclosing massif. A particular consequence of this statement is the well known condition of stick-slip occurrence.

It means that for episodes of slow movements to occur it is not obligatory that the fault is in a transitional state from brittle to plastic, as it happens either at large depths (25–45 km) between the seismogenic zone (beneath the zone) and the zone of stable sliding in subduction zones, where slow slip events are observed most often, or at shallow depths (~5 km) between the seismogenic zone (above the zone) and the surface zone of continuous creep [43]. Presence of watered clays in the principal slip zone, or of some amount of talc, which often substitutes the minerals of serpentite group along fracture walls in chemical reaction of serpentite with the silicon dioxide contained in thermal fluids, decreases harshly the shear stiffness of the fault, so that its value can be essentially lower than 10% of the normal one. In this situation the effective gradient of shear strength of the fault may turn to be close to the stiffness of the massif, which, can lead to occurrence of slow movements on faults. That is why similar effects can be observed at all depths in the crust.

As far as the possibility of artificial change of the slip mode on a fault is considered, the aim of external action should be not the removal of excessive stresses, but the decrease of fault zone stiffness. The change of sliding conditions should involve a

rather big area, appreciably bigger than the size of the zone of earthquake nucleation. For example, pumping a clay-containing suspension into a fault zone may provide such a result, but this difficult scientific and engineering problem demands the development of a detailed technique of fulfilling the operation and methods to estimate its consequences.

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Multilayer Modelling of Lubricated Contacts: A New Approach Based on a Potential Field Description



Markus Scholle, Marcel Mellmann, Philip H. Gaskell, Lena Westerkamp, and Florian Marner

Abstract A first integral approach, derived in an analogous fashion to Maxwell's use of potential fields, is employed to investigate the flow characteristics, with a view to minimising friction, of shear-driven fluid motion between rigid surfaces in parallel alignment as a model for a lubricated joint, whether naturally occurring or engineered replacement. For a viscous bilayer arrangement comprised of immiscible liquids, it is shown how the flow and the shear stress along the separating interface is influenced by the mean thickness of the layers and the ratio of their respective viscosities. Considered in addition, is how the method can be extended for application to the more challenging problem of when one, or both, of the layers is a viscoelastic material.

Keywords Lubrication theory · Finite elements · Complex variable analysis · Shear flow · Immiscible liquids · Viscoelasticity · Surface contouring · Joints · Bearings

1 Introduction and Model Assumptions

Hydrodynamic lubrication [1, 2] is one of the classic topics contributing to the field of fluid mechanics that is of considerable relevance; for example, in technological terms in connection with the design of lubricated contacts such as plain bearings or ball joints [3] and more specifically, in bio-engineering terms, in the context of joint replacement [4]. In the present work, a model, based on a potential field description is

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presented for the case of a lubricated contact consisting of two contiguous immiscible layers, located one on top of the other. In a general sense, both layers can be considered as purely viscous liquids and/or viscoelastic layers, depending on the application of interest.

Different aspects of film flows involving two or more immiscible liquid layers have been investigated in [5–7] with a focus on both confined shear-driven flows, the topic of interest here, and free-surface flows due to their relevance in the production and deposition of functional coatings [8]. Current studies addressing the material modelling of articular cartilage, see for example [9], reveal an appreciable complexity of material behaviour, which among other things includes chemo-elastic effects and anisotropy; here standard simplified viscoelastic models are considered as a first step.

The model problem considered is that of an idealised system of two-dimensional steady Couette flow, as illustrated in Fig. 1: the lower, flat surface translates with speed v_0 while the upper, corrugated, one remains stationary. The region separating the surfaces, which are in parallel alignment, is taken to be filled with contiguously contacting, immiscible liquids or viscoelastic layers, having different dynamic viscosity η and Young's modulus E ; the case shown is for a bilayer system, mimicking the more general case of a joint in which the synovia meets a protective layer exhibiting viscoelastic properties. While at outset the general case is formulated, the focus of the results presented and discussed subsequently is restricted to the simpler case of two Newtonian liquid layers.

The two-phase system is defined in terms of a number of non-dimensional parameters, the most relevant of these being the ratios of the two layer thicknesses, H_1/H_2 , and of their fluid properties, namely the viscosities η_1/η_2 and the densities ρ_1/ρ_2 . The shape of the periodic profile defining the upper corrugated surface is given by the function:

$$b(x) = -2a \frac{\ln(2 - 2s \cos x) - \ln(1 + \sqrt{1 - s^2})}{\ln(1 + s) - \ln(1 - s)}. \quad (1)$$

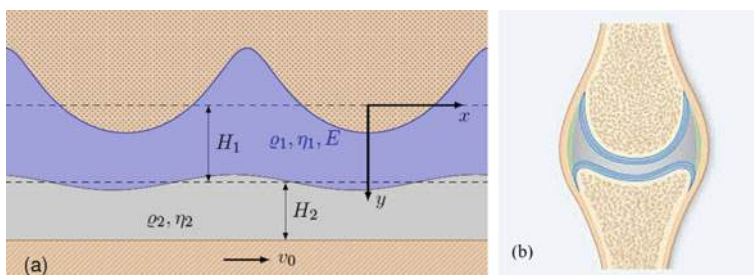


Fig. 1 Model of a periodic two-phase system (a), consisting of a layer of viscous liquid lying on top of a viscoelastic one, both confined between non-compliant rigid surfaces; the lower one is flat and translating while the upper one is profiled/contoured and stationary. The model is based on the natural form of biological joints (b), [10] and is a key feature of the planned investigation

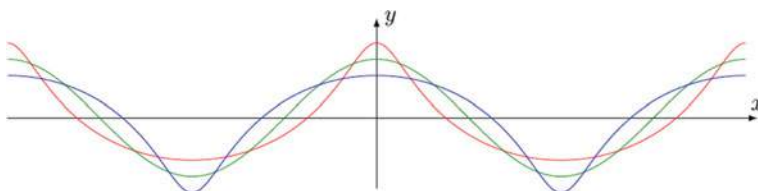


Fig. 2 Surface profile shapes obtained for different values of the shape parameters. The red curve, with $s = 0.9$, results in peak asperities while the blue one, with $s = -0.9$, (phase shift π) leads to smoother ones. The cosine function (green curve) is shown as a reference for the limit $s \rightarrow 0$

It depends on two parameters: the dimensionless amplitude $a = 2\pi A/\lambda$ and $s \in (-1, 1)$ determining the shape. Figure 2 illustrates three potential shapes, demonstrating the role of the shape parameter s . In the limiting case $s \rightarrow 0$ the surface shape becomes a cosine function, $b(x) = -a \cos x$, for positive values of s the corrugations form pronounced peak asperities while for negative values they result in a smoother levelling.

If the upper layer is assumed to be viscoelastic, the Deborah number $De = \eta_1 v_0 / E\lambda$ enters the problem as an additional parameter, while the Reynolds number is so small that it can be taken to be zero.

The focus here, from a bio-engineering viewpoint, is the determination of the normal and shear stresses along the interface separating the two layers: normal stresses, especially when periodically varying with time, have a positive influence on the nutrient supply to the articular cartilage by promoting the exchange of substances between the nutrient-containing synovial fluid and the partially porous articular cartilage; shear stresses, on the other hand, cause wear and thus have a detrimental effect [4].

2 Mathematical Formulation

The field equations for the different layer types, together with the boundary and interface conditions, are formulated below, making use of the first integral approach [11–13]. The benefits of the latter are: (i) an elegant implementation for arbitrary rheological models; (ii) a beneficial form for the dynamic condition at the interface separating the two layers.

2.1 Field Equations for Newtonian Layer Types

When one, or both layers, is assumed to be an incompressible viscous Newtonian liquid, resolving the associated flow requires a solution of the governing Navier-Stokes equations and accompanying continuity equation:

$$\rho(\vec{v} \cdot \nabla)\vec{v} = -\nabla p + \eta \nabla^2 \vec{v}, \quad (2)$$

$$\nabla \cdot \vec{v} = 0, \quad (3)$$

respectively, to obtain the velocity $\vec{v} = v_x(x, y)\vec{e}_x + v_y(x, y)\vec{e}_y$ and scalar pressure $p = p(x, y)$ fields. Defining a complex coordinate and complex velocity as:

$$\xi = x + iy, \quad (4)$$

$$v = v_x + iv_y, \quad (5)$$

and introducing the scalar potential Φ as an auxiliary unknown, facilitates integration of Eq. (2), leading finally to the following two complex field equations [14]:

$$\frac{\rho}{2}v^2 = 2\eta\frac{\partial v}{\partial \bar{\xi}} - 4\frac{\partial^2 \Phi}{\partial \bar{\xi}^2}, \quad (6)$$

$$\frac{\rho}{2}\bar{v}v = -p + 4\frac{\partial^2 \Phi}{\partial \bar{\xi} \partial \xi}. \quad (7)$$

The continuity equation is fulfilled identically on introduction of the streamfunction Ψ , according to $v = -2i\partial\Psi/\partial\bar{\xi}$.

Note that the above approach originates from two-dimensional elasticity theory [15, 16], where the scalar potential Φ plays the role of Airy's stress function. This prominent complex variable approach was adopted subsequently by several authors, e.g. [17], for the solution of Stokes flow problems, before being generalised for the case of inertial flows in [11].

2.2 Field Equations for Viscoelastic Layer Types

A complex variable formulation of the governing evolution equations, as in the case of Newtonian liquids and Hookean materials, is also available for any generalised material [18] with an elegant embodiment of the respective rheological model equations in the first integral approach. By applying the complex transformations (4) and (5) to the general momentum balance in place of the Navier-Stokes equations and following the procedure described in [14], one obtains the following complex equations:

$$\frac{\rho}{2}v^2 = \underline{\sigma} - 4\frac{\partial^2 \Phi}{\partial \bar{\xi}^2}, \quad (8)$$

Table 1 Substitution rules for the implementation of generalised rheological models

Rheological models	σ	$\dot{\sigma}$	E	$\dot{\varepsilon}$
w.r.t. Eq. (8)	$\underline{\sigma}$	$v \frac{\partial \underline{\sigma}}{\partial \xi} + \bar{v} \frac{\partial \underline{\sigma}}{\partial \bar{\xi}}$	$2 \frac{\partial u}{\partial \xi}$	$2 \frac{\partial}{\partial \xi}$
w.r.t. Eq. (9)	σ_0	$v \frac{\partial \sigma_0}{\partial \xi} + \bar{v} \frac{\partial \sigma_0}{\partial \bar{\xi}}$	$\frac{\partial u}{\partial \xi} + \frac{\partial \bar{u}}{\partial \bar{\xi}}$	$\frac{\partial v}{\partial \xi} + \frac{\partial \bar{v}}{\partial \bar{\xi}}$

$$\frac{\rho}{2} \bar{v} v = \sigma_0 + 4 \frac{\partial^2 \Phi}{\partial \xi \partial \bar{\xi}},$$

(9)

where $\sigma_0 = \frac{\sigma_x + \sigma_y}{2}$ is the isotropic part of the stress tensor of the respective material, while the complex quantity $\underline{\sigma} = \frac{\sigma_x - \sigma_y}{2} + i \tau_{xy}$ is its traceless part. The adoption of a corresponding rheological model, given as a relationship between the stress σ , the deformation ε , together with their time derivatives, can be implemented by formal substitutions according to the rules listed in Table 1.

Here $u = u_x + i u_y$ denotes the complex displacement field. Note also, the kinematic constraint $v = v \frac{\partial u}{\partial \xi} + \bar{v} \frac{\partial u}{\partial \bar{\xi}}$ between the velocity and displacement fields.

Implementation of the above methodology is demonstrated for a Kelvin-Voight model $\sigma = E\varepsilon + \eta\dot{\varepsilon}$ with Young's modulus E and viscosity η [19]. Following substitution, according to Table 1, Eqs. (8) and (9) become:

$$\frac{\rho}{2} v^2 = 2E \frac{\partial u}{\partial \xi} + 2\eta \frac{\partial v}{\partial \xi} - 4 \frac{\partial^2 \Phi}{\partial \xi^2},$$

(10)

$$\frac{\rho}{2} \bar{v} v = E \left(\frac{\partial u}{\partial \xi} + \frac{\partial \bar{u}}{\partial \bar{\xi}} \right) + \eta \left(\frac{\partial v}{\partial \xi} + \frac{\partial \bar{v}}{\partial \bar{\xi}} \right) + 4 \frac{\partial^2 \Phi}{\partial \xi \partial \bar{\xi}},$$

(11)

which are generalised forms of Eqs. (6) and (7), respectively; the latter equations result in the limit case of a viscous liquid when $E = 0$.

Note that the above procedure can be applied to any arbitrary rheological model, using the formal substitution rules in Table 1 with respect to the general complex momentum Eqs. (8) and (9).

2.3 Boundary and Interface Conditions

Along both the stationary profiled surface, given by the function $b(x)$, and the moving flat surface no-slip/no-penetration conditions have to be fulfilled:

$$u(x, b(x)) = 0,$$

(12)

$$v(x, H_1 + H_2) = v_0.$$

(13)

Periodic boundary conditions at inflow and outflow, to the left and right, are enforced. At the interface separating the layers, $y = f(x)$, the shape of which is unknown a priori, the velocity field has to be continuous:

$$[[v]] = 0, \quad (14)$$

with the double square brackets denoting the discontinuity of the associated term. Moreover, the kinematic boundary condition there:

$$v_y - f'(x)v_x = 0, \quad (15)$$

can be used to determine the shape of the interface. Finally, the dynamic interface condition:

$$\left[\underline{\underline{T}}_2 - \underline{\underline{T}}_1 \right] \cdot \vec{n} = \sigma_S \kappa \vec{n}, \quad (16)$$

accounts for the equality in stress at the interface; σ_S is the interfacial tension, κ the curvature, \vec{n} the vector normal to the interface and $\underline{\underline{T}}_{1,2}$ the stress tensor associated with the materials forming the respective layers. Using a conventional description, the treatment of the dynamic interface condition is a challenging task, since the viscous or viscoelastic stresses present lead to combinations of different derivatives of different components of the displacement and velocity field and therefore to a mathematically unfavourable form. Using the first integral approach, the unfavourable terms involved in the interface condition (16) can be replaced by second order derivatives of the scalar potential Φ and the interface condition integrated [14], leading finally to the simple jump condition:

$$\left[\left[\frac{\partial \Phi}{\partial \bar{\xi}} \right] \right] = \frac{\sigma_S n}{4}, \quad (17)$$

where $n = n_x + i n_y$ is the complex equivalent of the normal vector. It is shown in [14] that, after re-transformation to a real-valued representation, the two conditions resulting from (17) can be formulated in standard Dirichlet/Neumann form. Among various other benefits, the reduction of the complicated dynamic interface condition (16) to the significantly simpler jump condition (17) for the potential Φ justifies its introduction as an additional auxiliary field and demonstrates its use.

3 Methods of Solution

The problem is formulated in three different ways: via a Lubrication approximation allowing for an analytical solution; a numerical finite-element FE approach; a semi-analytical one benefitting from the use of complex variables. The FE approach

enables Reynolds number effects to be investigated and, more generally, the validity of the two simpler models to be assessed. In terms of the adopted first integral approach, the standard mathematical form of the jump conditions (17) is advantageous, since the resulting friction coefficient can be calculated conveniently from the auxiliary potential field Φ based on the first integral formulation without the need to approximate velocity derivatives in a post-processing step as would be the case if a primitive variable formulation had been adopted.

3.1 Lubrication Approximation

The first integral Eq. (6) can be simplified based on a Lubrication approximation, noting that the same applies to the Navier-Stokes equations, [20–23], leading to a single equation for the local film thickness; the velocity field to leading order being locally parabolic. The requirement underpinning its applicability in the case of surface contours exhibiting rapid changes is that the commensurate interface disturbance is slowly varying.

Applying the Lubrication approximation to the real-valued decomposed Eq. (6) leads to:

$$\begin{aligned} \frac{\rho}{2}(v_x^2 - v_y^2) + \frac{\partial}{\partial x} \left[\frac{\partial \Phi}{\partial x} - \eta v_x \right] - \frac{\partial}{\partial y} \left[\frac{\partial \Phi}{\partial y} - \eta v_y \right] &= 0, \\ \rho v_x v_y + \frac{\partial}{\partial y} \left[\frac{\partial \Phi}{\partial x} - \eta v_x \right] + \frac{\partial}{\partial x} \left[\frac{\partial \Phi}{\partial y} - \eta v_y \right] &= 0. \end{aligned} \quad (18)$$

Next, by neglecting $\partial v_y / \partial x$ compared to $\partial v_x / \partial y$ and omitting inertial terms, the set of PDEs is reduced to one that can be solved by successive integration, as shown in detail in [24], leading to the following general solution:

$$v_x = \frac{1}{2} F_1''(x) y^2 + F_2'(x) y + F_3(x), \quad (19)$$

$$\psi = F_1''(x) \frac{y^3}{6} + F_2'(x) \frac{y^2}{2} + F_3(x) y + F_4(x), \quad (20)$$

for the velocity v_x and the streamfunction ψ , involving four integration functions $F_1(x)$, $F_2(x)$, $F_3(x)$ and $F_4(x)$; while the gradient of the potential results in:

$$\begin{aligned} \frac{2}{\eta} \frac{\partial \Phi}{\partial x} &= \frac{1}{2} F_1''(x) y^2 + F_2'(x) y + F_3(x) + F_1(x), \\ \frac{2}{\eta} \frac{\partial \Phi}{\partial y} &\approx F_1'(x) y + F_2(x). \end{aligned} \quad (21)$$

Note that four integration functions occur with respect to each layer m , thus for a bilayer system eight functions $F_{mi}(x)$ with $m = 1, 2$ and $i = 1, \dots, 4$ have to be considered. Together with the shape of the interface, $f(x)$, nine unknown functions have to be determined from the boundary and interface conditions (12)–(15) and (17), considering that each of the complex conditions delivers two real-valued conditions after decomposition into real and imaginary parts.

By successive elimination of unknown functions, the resulting set of nine ODEs can be reduced to three nonlinear ODEs for the functions $F_{11}(x)$, $F_{21}(x)$ and $f(x)$. In non-dimensional form, taking $L = \lambda/2\pi$ as the characteristic length with λ the wavelength and v_0 as a characteristic velocity, the resulting equations read:

$$\frac{2Q_2}{h-f(x)} - \frac{2Q_1}{f(x)-b(x)} - \frac{[f(x)-b(x)]^2}{6}F''_{11}(x) + \frac{[h-f(x)]^2}{6}F''_{21}(x) = 1 \quad (22)$$

$$(1-n)\left[\frac{2Q_1}{f(x)-b(x)} + \frac{[f(x)-b(x)]^2}{6}F''_{11}(x)\right] + F_{21}(x) - nF_{11}(x) = 0 \quad (23)$$

$$\begin{aligned} & \frac{Q_2}{[h-f(x)]^2} + \frac{nQ_1}{[f(x)-b(x)]^2} + n\frac{f(x)-b(x)}{3}F''_{11}(x) \\ & + \frac{h-f(x)}{3}F''_{21}(x) = \frac{1}{h-f(x)} \end{aligned} \quad (24)$$

where $Q_1 = \psi(x, f(x))$ and $Q_2 = \psi(x, h) - Q_1$ are the constant partial flow rates of the two different layers and $n = \eta_1/\eta_2$ is the viscosity ratio. In contrast to the conventional Lubrication model derived starting from the original Navier-Stokes equations leading to the well-known Reynold's equation [3], the first integral approach yields the equation set (22)–(24) comprising three equations for the interface shape $f(x)$ and the two functions $F_{11}(x)$, $F_{21}(x)$ which are connected with the curvatures of the respective velocity profiles within the two layers. Having once determined these three functions by solving (22), (23) and (24), the streamfunction results for the two layers, as:

$$\begin{aligned} \psi^{(1)} &= \frac{[y-b(x)]^2}{2} \left[\frac{y-f(x)}{3}F''_{11}(x) + \frac{2Q_2}{[f(x)-b(x)]^2} \right], \\ \psi^{(2)} &= Q_1 + \left[1 - \frac{(h-y)^2}{[h-f(x)]^2} \right] [Q_2 - h + f(x)] \\ &+ \frac{(h-y)^2}{6} [y-f(x)]F''_{21}(x), \end{aligned} \quad (25)$$

where the numbers 1, 2 in the brackets denote the respective layer.

The above set of nonlinear equations, (22), (23) and (24), can be solved numerically in their given form or asymptotically after linearization, as shown below. The latter approach is briefly demonstrated in the following. By introducing:

$$\varphi(x) := f(x) - h_1, \quad (26)$$

as the deviation of the interface from its mean value and assuming that the functions $b(x)$, $\varphi(x)$, $F''_{11}(x)$ and $F''_{21}(x)$ depend linearly on the amplitude a of the profiled upper surface, an asymptotic expansion of Eqs. (22) and (24) with respect to powers of a leads, to zeroth order, to:

$$\begin{aligned} Q_1 &= \frac{h_1}{h_1 + nh_2} \frac{h_1}{2}, \\ Q_2 &= \frac{2h_1 + nh_2}{h_1 + nh_2} \frac{h_2}{2}, \end{aligned} \quad (27)$$

as solutions for the flow rates in the case of bilayer flow between two parallel planar walls. The first order contribution of the same equations resulting from linearization with respect to a :

$$\begin{aligned} \frac{h_1}{3}(h_1 + nh_2)F''_{11}(x) &= \frac{2h_1 + nh_2}{h_1(h_1 + nh_2)}[\varphi(x) - b(x)] + \frac{3h_1 + 2nh_2}{h_2(h_1 + nh_2)}\varphi(x), \\ \frac{h_2}{3}(h_1 + nh_2)F''_{21}(x) &= -\frac{n[\varphi(x) - b(x)]}{h_1 + nh_2} - \frac{h_1^2 + 4nh_1h_2 + 2n^2h_2^2}{h_2^2(h_1 + nh_2)}\varphi(x), \end{aligned} \quad (28)$$

allow the two functions $F''_{11}(x)$ and $F''_{21}(x)$ to be expressed in terms of $\varphi(x)$ and $b(x)$. By computing the linear part of (23), taking the second order derivative with respect to x and eliminating $F''_{11}(x)$ and $F''_{21}(x)$ by means of (28), one ends up with a second order ODE of the form:

$$A_1\varphi''(x) + A_2\varphi(x) = A_3b''(x) + A_4b(x), \quad (29)$$

with the four coefficients given by:

$$\begin{aligned} A_1 &= -(1-n)(3h_1^2 + 2nh_1h_2 - nh_2^2) \\ A_2 &= 6\left(\frac{h_1}{h_2} + \frac{nh_2}{h_1}\right)^2 + 12n\left[2\frac{h_1}{h_2} + (1+n)\left(1 + \frac{h_2}{h_1}\right)\right] \\ A_3 &= n(1-n)h_2^2 \\ A_4 &= 6n\left[1 + 2\frac{h_2}{h_1} + n\frac{h_2^2}{h_1^2}\right] \end{aligned} \quad (30)$$

Since the above ODE (29) is linear and of second order, it allows for a closed form analytic solution for any prescribed profile shape $b(x)$. Note that the latter need not to be periodic at this stage; apart from the examples considered in the results section, any profile shape can be considered, including step and trench geometries as in [24].

3.2 *Finite Elements Approach*

For validation purposes numerical calculations are obtained, in the case of two viscous layers, by using existing and established practices based on classical FE formulations for the original Eqs. (2) and (3) in terms of “primitive variables”, namely velocity and pressure, see e.g. [25, 26] or [27]; an alternative approach would have been to use a streamfunction and vorticity formulation as adopted by [28, 29] or [30].

A challenging task is the a priori unknown shape $f(x)$ of the interface separating the two layers, requiring an iterative approach in which the calculation for the two system components, no matter whether they are fluid or viscoelastic, is carried out separately, assuming a starting value for $f(x)$ and calculating the velocity/displacement field for both layers without considering the kinematic boundary condition (15). Following this, a new interface shape $f(x)$ is calculated separately as the limiting streamline. The iteration process is repeated until either the change in the interface shape from one iteration step to the next falls below a prescribed tolerance or if the ratio of the normal velocity to the tangential one along the current interface shape is smaller than a tolerable value, typically 0.25% [31]. Implementation of the methodology was performed using standard libraries for efficient FE Galerkin solvers, making use of the packages ‘numpy’, ‘scipy’ and ‘matplotlib’, accessed via Python and the ‘Triangle’ mesh generator [32].

3.3 *Complex-Variable Approach with Spectral Solution Method*

For completeness and different to the above aforementioned approaches, direct use can be made of the complex formulation (10) and (11) of the field equations. On neglecting the nonlinear inertial terms, Eq. (10) becomes integrable with respect to $\bar{\xi}$, implying:

$$Eu + \eta v = 2 \frac{\partial \Phi}{\partial \bar{\xi}} - 4g_0(\xi), \quad (31)$$

containing the integration function $g_0(\xi)$. After inserting this into Eq. (11), the identity:

$$g'_0(\xi) + \overline{g'_0(\xi)} = \frac{\partial^2 \Phi}{\partial \bar{\xi} \partial \xi}, \quad (32)$$

is obtained, which is again integrable. Further integration and noting that the potential Φ is real-valued, finally leads to:

$$\Phi = 2\Re[\bar{\xi}g_0(\xi) + g_1(\xi)], \quad (33)$$

yielding a second integration function $g_1(\xi)$. Thus, the entire problem has been reduced to one of determining the two holomorphic functions $g_0(\xi)$ and $g_1(\xi)$, frequently referred to as Goursat functions.

For the incompressible flow of a Newtonian liquid $v = -2i\partial\psi/\partial\bar{\xi}$ and $E = 0$, in which case Eq. (31) can be integrated a second time leading, together with expression (33), to the advantageous form:

$$\Phi + i\eta\psi = \bar{\xi}g_0(\xi) + g_1(\xi), \quad (34)$$

of the solution for both the streamfunction and the scalar potential [12].

Since the Goursat functions are functions of only one complex variable, the mathematical problem is elegantly reduced from a two- to a one-dimensional problem. While in the classical literature a purely analytical approach via conformal mappings is preferred, which is limited to finding approximate solutions for simple geometries, a spectral method based on a Fourier expansion of either the Goursat functions directly or their boundary values enables fulfillment of the boundary conditions for arbitrary profile shapes with arbitrary accuracy, depending of the truncation order of the Fourier series. Although not considered in the context of the result presented and discussed below, further details of the use of this elegant semi-analytical method and its application to free surface flows and Couette flows for Newtonian liquids can be found in [31, 33, 34].

4 Results

In the present work bilayer flow is explored, for a stationary upper surface having a particular contoured shape, utilising the Lubrication approach with the results validated via corresponding FE-calculations.

4.1 Sinusoidal Upper Surface Shapes

Assuming a sinusoidally shaped profile for the upper stationary surface, $b(x) = -a \cos x$, resulting as the limit case $s \rightarrow 0$ of the more general shape (1), the solution of the ODE (29) is obtained straightforwardly by assuming a corresponding harmonic form for the interface shape, i.e.: $\varphi(x) = -\hat{\varphi}a \cos x$, with the amplitude factor $\hat{\varphi}$ resulting in:

$$\hat{\varphi} = \frac{A_4 - A_3}{A_2 - A_1}, \quad (35)$$

and with the coefficients A_1, \dots, A_4 given according to (30). From the above solution the interface function $f(x)$ and the two functions $F_{11}(x)$, $F_{21}(x)$ are obtained from (26) and (28) and finally the streamfunction from (25).

This closed form analytic solution for a viscous bilayer flow is visualised via streamlines in the left column of Fig. 3 for a fixed thickness ratio of $h_1/h_2 = 2/3$ and a fixed amplitude $a = 1/2$ for varying viscosity ratio n . Corresponding FE solutions are provided in the right column.

The calculations reveal a clear dependence of the interface shape on the viscosity ratio n : if the viscosity η_1 of the layer adjacent to the profiled surface is much larger than the viscosity η_2 of the layer adjacent to the planar translating one, the interface shape mimics the upper surface profile. Comparing the resulting flow for $n = 2$ with the corresponding one for $n = 500$, a minimal change only is apparent, indicating that for a very large ratio a limit case exists. If, vice versa, the viscosity of the lower layer is larger than the viscosity of the upper one, the interface becomes smoother. For n with a very small value, the second layer acts effectively as a continuation of the translating planar surface and the interface approaches a straight line. For the

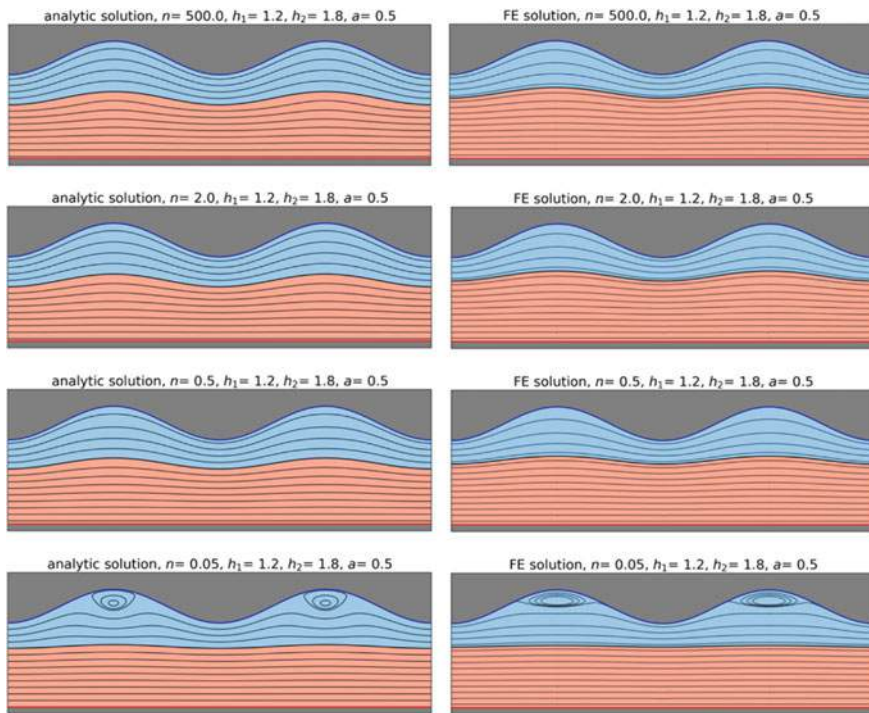


Fig. 3 Streamline plots of bilayer flow for the case of a sinusoidally profiled upper surface, with fixed geometry and varying viscosity ratio. The analytical results stem from the linearized Lubrication approximation (left column) and are compared to their corresponding FE solutions (right column)

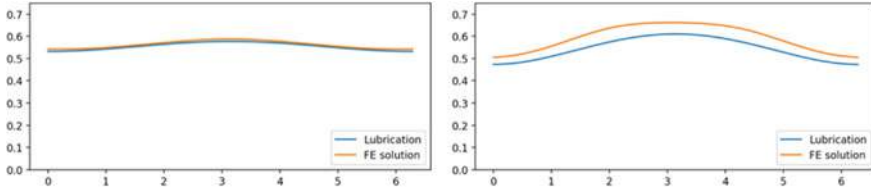


Fig. 4 Resulting shear stress along the interface of a bilayer flow over one wavelength; the layer thicknesses are $h_1 = 1.2$, $h_2 = 1.8$ and $n = 1/3$, for two different amplitudes, $a = 0.1$ (left) and $a = 0.3$ (right), calculated via the Lubrication and FE approaches

present geometry this induces the onset of a small eddy in the troughs of the profiled surface, which is a well-known observation in monolayer Couette flow, see e.g. [35], or [36].

The above results show that the solution obtained via an asymptotic analysis, although slightly overestimating the slope of the interface shape, leads to a good approximation for bilayer flow in the presence of a sinusoidally profiled surface; its limitations are due primarily to the prerequisite of a small corrugation amplitude a . Additionally, one has to keep in mind that the lubrication analysis is effectively a long-wave approximation [21] requiring the film thickness not to exceed the wavelength of the upper surface profile.

As mentioned in the introduction, the shear stress τ_f along the interface $y = f(x)$ is of particular interest and can, in general, be calculated as:

$$\tau_f = 2\eta \left[\frac{\partial v}{\partial y} - \frac{\partial u}{\partial x} \right] \frac{f'(x)}{1 + f'(x)^2} + \eta \left[\frac{\partial u}{\partial y} + \frac{\partial v}{\partial x} \right] \frac{1 - f'(x)^2}{1 + f'(x)^2}. \quad (36)$$

Two examples of the resulting distribution of the shear stress along the interface over one wavelength of the bilayer flow are presented in Fig. 4, for the same layer thicknesses $h_1 = 1.2$ and $h_2 = 1.8$ considered when generating the streamline patterns for the flows in Fig. 3 but with a viscosity ratio $n = 1/3$ and for two different profile amplitudes.

As expected, the maximum shear stress coincides with the peak value of the upper surface profile where the local film thickness is a minimum. Furthermore, for the smaller of the two amplitudes, $a = 0.1$, the agreement between the analytically calculated shear stress and the comparative FE solution is excellent; while for the larger amplitude, $a = 0.3$, the shear stress is underestimated by the Lubrication approach.

4.2 Inharmonic Periodic Upper Surface Profiles

Due to the linearity of the ODE (29), the result obtained for sinusoidally profiled upper surfaces can be adapted to generally periodic profiles by utilizing a spectral

decomposition:

$$b(x) = a \sum_{k=1}^{\infty} \beta_k \cos(kx), \quad (37)$$

of the respective profile shape function. For the shape given by Eq. (1), the Fourier coefficients read [37]:

$$\beta_k = \frac{4(1 - \sqrt{1 - s^2})^k}{k[\ln(1 + s) - \ln(1 - s)]s^k}. \quad (38)$$

This allows the ODE (29) to be solved separately for each spectral component and the writing of the solution for the interface shape as the superposition:

$$\varphi(x) = a \sum_{k=1}^{\infty} \varphi_k \beta_k \cos(kx), \quad (39)$$

where:

$$\varphi_k = \frac{A_4 - k^2 A_3}{A_2 - k^2 A_1}. \quad (40)$$

FE solutions were generated in the same manner as above. Figure 5 shows the streamline patterns obtained for a bilayer flow in the presence of an upper surface profile given by the analytic form (1), with $a = 0.5$ and a shape parameter $s = 0.9$ and as before layer thickness of $h_1 = 1.2$ and $h_2 = 1.8$, for three different values of n . The corresponding interface shape given by the Lubrication approximation is shown as a dashed line in each case.

As can be seen the results obtained are qualitatively similar to those of Fig. 3 for the case of a sinusoidally shaped upper surface profile: for the larger of the three n values the interface disturbance is greater, while for the smaller of the two n values the interface tends to a straight line when $n = 0.05$, and for which case distinct eddies are observed to exist in the troughs of the profiled surface. As for the case of a sinusoidally profiled upper surface, it can be seen that the curvature of the interface is overestimated by the Lubrication approach.

5 Conclusions and Perspectives

Three different approaches are presented as solutions to the problem of bilayer flow for the case of two immiscible Newtonian liquids confined between an upper profiled

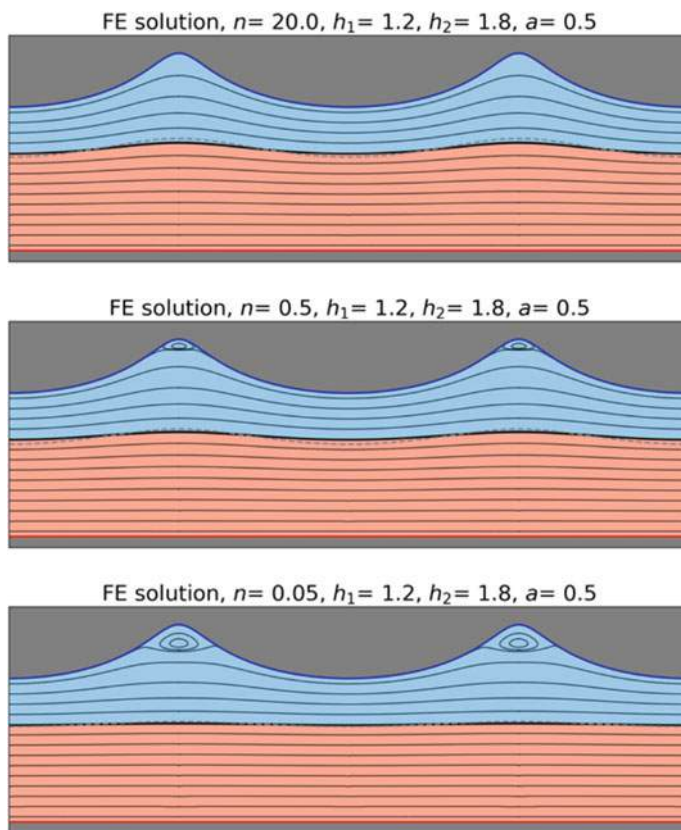


Fig. 5 Streamline plots for bilayer flow, obtained using the FE approach, in the presence of an inharmonic upper surface profile, with fixed geometry and three different viscosity ratios. The interface shape in each case, obtained analytically via the Lubrication approximation with linearization, is shown for comparison purposes as a dashed line

surface at rest and a lower translating planar surface. The FE approach enables sufficiently accurate solutions of the Navier-Stokes equations to be obtained, that provide a reliable means of validating the predictions of the other two methods. The latter originate from a potential-based first integral formulation of Navier-Stokes equation. In the present work only the Lubrication approach in combination with linearization of the resulting ODEs is considered. This allows for closed form analytic solutions, which are compared in detail with corresponding FE solutions. Although overestimating the curvature of the interface between the two fluid layers and underestimating the shear stress along the latter, the Lubrication approach provides results of quantitatively acceptable accuracy if the amplitude of the profiled upper surface is sufficiently small. A potential improvement of the method could be realised by solving the nonlinear Eqs. (22), (23) and (24) without linearization.

For both biomedical and advanced technological applications, the consideration of non-Newtonian materials is an essential next step. In this context the complex variable approach outlined above provides an interesting perspective towards the implementation of viscoelastic models within a first integral framework; further details of which will appear in forthcoming articles.

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Microstructure-Based Computational Analysis of Deformation and Fracture in Composite and Coated Materials Across Multiple Spatial Scales



Ruslan R. Balokhonov and Varvara A. Romanova

Abstract A multiscale analysis is performed to investigate deformation and fracture in the aluminum-alumina composite and steel with a boride coating as an example. Model microstructure of the composite materials with irregular geometry of the matrix-particle and substrate-coating interfaces correspondent to the experimentally observed microstructure is taken into account explicitly as initial conditions of the boundary value problem that allows introducing multiple spatial scales. The problem in a plane strain formulation is solved numerically by the finite-difference method. Physically-based constitutive models are developed to describe isotropic strain hardening, strain rate and temperature effects, Luders band propagation and jerky flow, and fracture. Local regions experiencing bulk tension are found to occur during compression that control cracking of composites. Interrelated plastic strain localization in the steel substrate and aluminum matrix and crack origination and growth in the ceramic coating and particles are shown to depend on the strain rate, particle size and arrangement, as well as on the loading direction: tension or compression.

Keywords Composites · Coated materials · Constitutive modeling · Plastic strain localization · Fracture · Multiscale numerical simulation

1 Introduction

Actual materials have essentially inhomogeneous microstructure (Fig. 1). According to the concepts of physical mesomechanics, stress concentrators of different physical origin are a major factor influencing the deformation pattern in nonhomogeneous materials. The effects are most conspicuous in composite materials (metal-matrix composites, coated and surface-hardened materials, doped alloys, etc.) because of differences in the mechanical properties (density, elastic moduli, and strength and plasticity characteristics) of their constituent elements. Thus, basic research along

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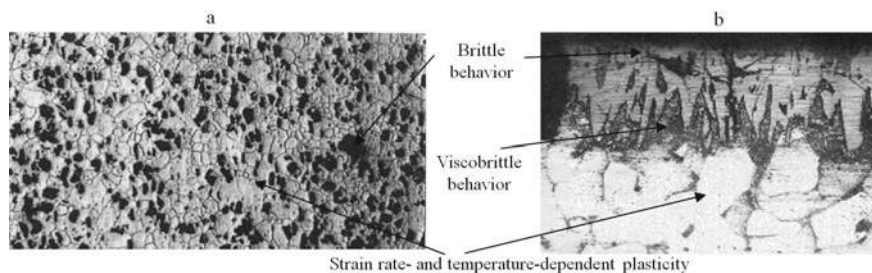


Fig. 1 Microstructures of an Al/Al₂O₃ composite (a) and steel coated by diffusion borating (b) [16, 34]

these lines is of great practical importance for development of advanced structural and functional materials.

The foundations of the physical mesomechanics of materials as a consistent methodology were laid more than 35 years ago [1]. At this time, the basic principles underlying the scientific approach have been formulated and developed [2, 3]. Early theoretical studies helped outline the range of immediate tasks and determined modeling and simulation techniques capable of solving these problems [2, 4–10]. New deformation and fracture mechanisms operative at macro-, meso- and microscales in solids were identified and accounted for [4]. Development of techniques for computer-aided design of materials and software enabled deformation and fracture processes to be simulated [2, 4–10].

At present, the problem of an adequate consideration of the multiscale nature of solids is recognized internationally as the first-priority line of investigations aimed at developing new-generation materials, and there is an increasing interest in theoretical studies in this research area—see e.g. [11–22]. This has been due to a general awareness that a correct prediction of the macroscopic properties of solids is hardly possible without hierarchy of structural levels and scales in the materials under study.

Nowadays, there are a large number of studies addressing multiscale numerical simulation and modeling, with an explicit consideration of the microstructure being taken into account. Material components are associated with proper constitutive models. For instance, some papers involve artificial models of real microstructures in studying micro-, meso- and homogenized macromechanical response of materials [2], [4, 9–11, 15, 17–19, 23–32]. Other authors reported results on the stress-strain analysis of experiment-based microstructure models [2, 6, 8, 12–14, 16, 20–22, 33–39]. All these and related works extend our understanding of the relationship between the microstructure and mechanical properties of materials. A special attention is given to interfaces. A majority of contributions devoted to the interfacial problem considers the interfacial fracture, decohesion and debonding [40–48]. Nevertheless, a comprehensive study of the phenomena related to the irregular interface geometry effects is often neglected.

The main purpose of this contribution is to show that, from the standpoint of mechanics stress concentration in local regions of a composite at different scale levels

could be of the same origin. It is controlled by the irregular geometry of microstructural elements making up the composition (ductile matrix/substrate, brittle ceramic particles/coatings/hardened interlayers, etc.) and the difference in their mechanical properties.

It is found out that the value of local stresses in composites might, by a large factor, exceed the average level of the load applied. The evolution of this effect is attributed to the presence of inhomogeneities with characteristic sizes corresponding to different scales—macro, meso and micro. It is demonstrated that the regions of stress concentration might undergo both compressive and tensile stresses irrespective of the type of external loading. The larger the difference in the mechanical properties of the constituents, the higher is the level of stress concentration developed in the vicinity of inhomogeneities of certain geometry.

The main aim of the paper is to investigate mechanisms of deformation and fracture which are related to complex geometry of interfaces in a composite material. Multi-scale analysis of deformation and fracture in composites is performed. A dynamic boundary-value problem is solved numerically by the finite-difference method. Constitutive models for the elastic-plastic deformation and elastic-brittle fracture are developed to describe the mechanical response of steel substrate/aluminum matrix and boride coating/alumina particles. Interface geometries correspond to the configurations found experimentally and are accounted for explicitly in the calculations.

2 Numerical Modelling Across Multiple Spatial Scales

In the frame of the proposed formulation, a multiscale numerical analysis implies at least two factors (Fig. 2): (1) the use of different models to describe the mechanical response of different constituent elements of a composite material under load in order to characterize the physical processes developing in the components and their

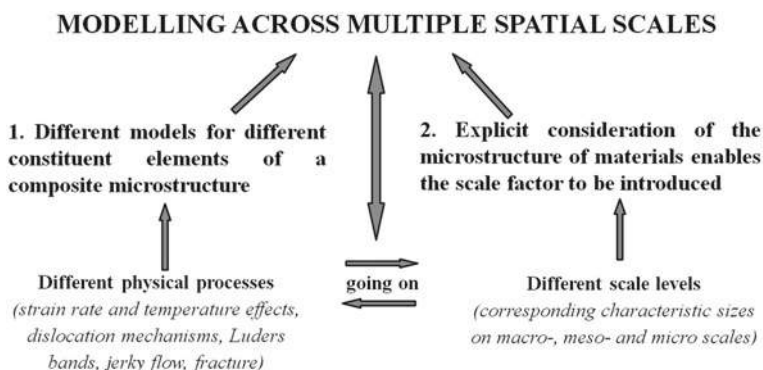


Fig. 2 Schematics of multiscale numerical simulation

interplay (Fig. 3), and (2) an explicit consideration of the microstructure of the material that provides information on the characteristic scales for which the models used are valid (Fig. 4).

It is suggested that deformation of composite materials can be described by a system of equations using the laws of conservation of mass and momentum, strain equations, and constitutive relations for the material constituents complemented with initial and boundary conditions (Fig. 3). The models presented in Fig. 3 enable us to handle only particular problems. Notably, the models under discussion have

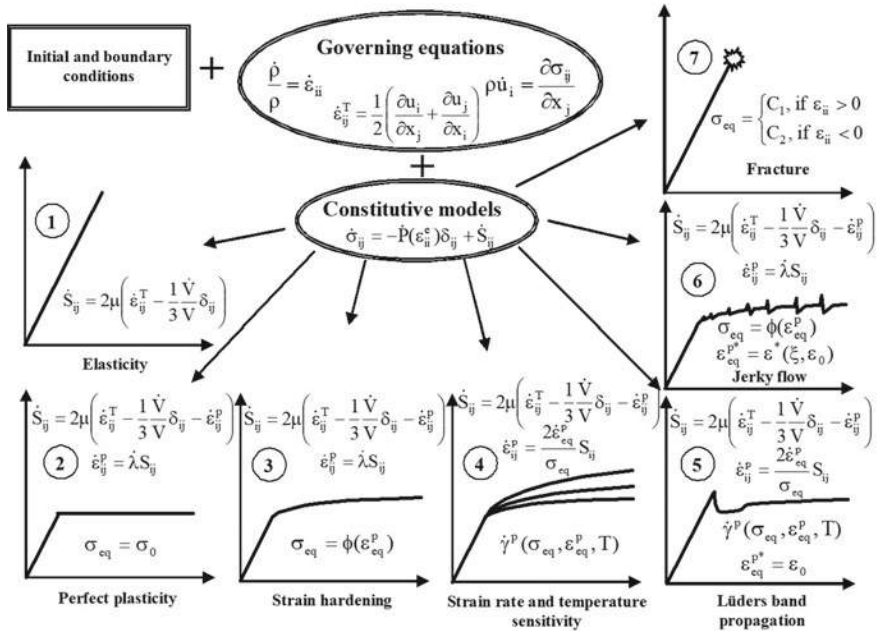


Fig. 3 A set of constitutive models for composite material constituents

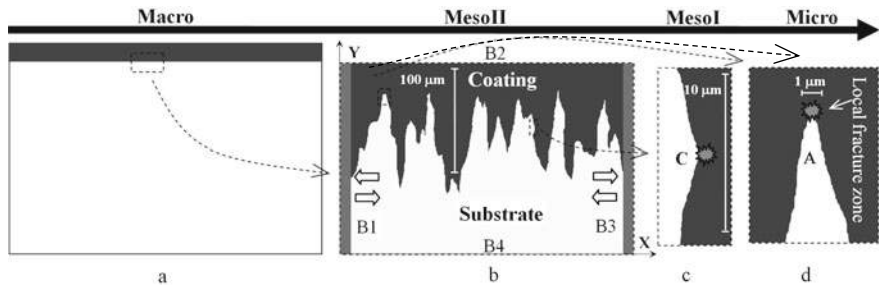


Fig. 4 Microstructure of steel with a hardened boride surface layer to be simulated and characteristic sizes of structure inhomogeneities at different scale levels

already been tested. The list of those accounting for the great diversity of the physical phenomena and processes observed in loaded solids is by no means complete.

The choice of particular elastoplastic response models numbered 1 through 6 in Fig. 3 depends on the material used as a substrate or a matrix (aluminum-based alloys, different types of steel, etc.) and on the applied loading conditions (high strain rate deformation, quasi-static loading, etc.). Fracture models, such as model 7, are required to describe the mechanical behavior of brittle and viscobrittle inclusions, coatings, and intermediate subsurface layers. Purely elastic or perfectly plastic descriptions (model 1 and 2) can be used as a first approximation.

An explicit consideration of the material microstructure makes it possible to introduce a scale factor and specify the length scales where one or another model can successfully be employed (Fig. 4). Figure 1b shows the mesostructure of a coated material, which was used in the calculations and corresponds to that observed experimentally (see Fig. 1). For the case in question, where the interface has a serrated profile and irregular geometry, we might single out certain types of inhomogeneities at different scale levels and their respective characteristic sizes. In particular, boride teeth proper, whose size is $\sim 50\text{--}100\text{ }\mu\text{m}$ (Fig. 4b), are independent stress concentrators. A quasiperiodic alternation of boride and steel teeth results in the formation of a peculiar stress-strained state at mesoscale II.

The shape of an individual tooth, is not perfect and exhibits fine structure at a lower scale level. Throughout the interface profile, there are convexities and concavities. The characteristic size of these inhomogeneities is within $5\text{--}10\text{ }\mu\text{m}$ (Fig. 4c). The regions of “intrusion” of ductile steel into a more brittle and strong boride material are sources of geometrical stress concentration at mesoscale I. Let us single out two types of such regions with respect to the direction of applied loading: types C and A, along (Fig. 4c) and perpendicular (Fig. 4d) to the x-direction.

The local fracture zone has a characteristic size of $\sim 1\text{ }\mu\text{m}$ (Fig. 4d) and gives rise to stress concentration at the microscale.

A homogenized stress-strain curve reflects the mechanical behavior of the mesovolume at the macroscale. Thus, an introduction of the mesovolume with a boride hardened layer of a complicated geometry in an explicit form allows us to prescribe the length scales—a scale hierarchy of inhomogeneities, whose characteristic sizes might differ by two orders of magnitude.

3 Governing Equations and Boundary Conditions

Let us formulate the governing equations (Fig. 3) in terms of plane strain. In this case there are the following non-zero components of the strain rate tensor:

$$\dot{\varepsilon}_{xx} = \dot{u}_{x,x}, \quad \dot{\varepsilon}_{yy} = \dot{u}_{y,y}, \quad \dot{\varepsilon}_{xy} = \frac{1}{2}(\dot{u}_{x,y} + \dot{u}_{y,x}), \quad (1)$$

where u_x and u_y are the components of the displacement vector, ε_{xx} , ε_{yy} and ε_{xy} are the strain tensor components, the upper dot and comma in the notations stand for the time and space derivatives, respectively.

The mass conservation law and the equations of motion take the forms

$$\dot{V}/V = \dot{\varepsilon}_{xx} + \dot{\varepsilon}_{yy}, \quad (2)$$

$$\sigma_{xx,x} + \sigma_{yx,y} = \rho \ddot{u}_x, \quad \sigma_{xy,x} + \sigma_{yy,y} = \rho \ddot{u}_y, \quad (3)$$

where σ_{xx} , σ_{yy} and σ_{xy} are the stress tensor components, V is the specific volume and ρ is the mass density.

Taking into account the resolution of the stress tensor in the spherical and deviatoric parts

$$\sigma_{ij} = -P\delta_{ij} + S_{ij} \quad (4)$$

the pressure and the stress deviator components are written as follow

$$\begin{aligned} \dot{S}_{xx} &= 2\mu \left(\dot{\varepsilon}_{xx} - \frac{1}{3} \dot{\varepsilon}_{kk} - \dot{\varepsilon}_{xx}^p \right), \quad \dot{S}_{yy} = 2\mu \left(\dot{\varepsilon}_{yy} - \frac{1}{3} \dot{\varepsilon}_{kk} - \dot{\varepsilon}_{yy}^p \right), \\ \dot{S}_{zz} &= 2\mu \left(-\frac{1}{3} \dot{\varepsilon}_{kk} \right) = -(\dot{S}_{xx} + \dot{S}_{yy}), \quad \dot{S}_{xy} = 2\mu (\dot{\varepsilon}_{xy} - \dot{\varepsilon}_{xy}^p), \quad \dot{P} = -K \dot{\varepsilon}_{kk}, \end{aligned} \quad (5)$$

where K and μ are the bulk and shear moduli, $\dot{\varepsilon}_{ij}^p$ is the plastic strain rate tensor, and δ_{ij} is the Kronecker delta.

To eliminate an increase in stress due to rigid rotations of medium elements, we define deviatoric stresses through the Jaumann derivative

$$\dot{S}_{ij}^* = \dot{S}_{ij} - S_{ik}\omega_{jk} - S_{jk}\omega_{ik}, \quad (6)$$

where $\omega_{ij} = \frac{1}{2}(\dot{u}_{i,j} - \dot{u}_{j,i})$ is the material spin tensor.

The strain tensor is the sum of elastic and plastic strain $\varepsilon_{ij} = \varepsilon_{ij}^e + \varepsilon_{ij}^p$, and $\dot{\varepsilon}_{kk}^p = 0$ is the hypothesis of plastic incompressibility. Unloading is elastic.

The boundary conditions on the surface B_1 and B_3 simulate uniaxial tension parallel to the X-axis, whereas on the bottom and top surfaces, they correspond to symmetry and free surface conditions, respectively (Fig. 4b). We obtain

$$\begin{aligned} \dot{u}_x &= \text{const} = -v \text{ for } (x, y) \in B_1, \quad \dot{u}_x = \text{const} = v \text{ for } (x, y) \in B_3, \quad \dot{u}_y \\ &= 0 \text{ for } (x, y) \in B_4, \quad \sigma_{ij} \cdot n_j = 0 \text{ for } (x, y) \in B_2, \quad \sigma_{xy} \\ &= 0 \text{ for } (x, y) \in B_1 \cup B_3 \cup B_4 \end{aligned} \quad (7)$$

Here $B = B_1 \cup B_2 \cup B_3 \cup B_4$ is the boundary of the computational domain, t is the computation time, v is the load velocity and n_j is the normal to the surface B_2 .

The system of Eqs. (1–7) have to be completed with a formulation for plastic strain rates $\dot{\varepsilon}_{ij}^p$.

4 Constitutive Modelling for Plasticity of the Substrate and Matrix Materials

4.1 Physically-Based Strain Hardening

Depending on the loading strain rate and material to be used as a substrate or matrix, different formulations of the constitutive models (models 2–6 in Fig. 3) have to be applied. To describe plasticity (model 3 in Fig. 3), use was made of the plastic flow law

$$\dot{\varepsilon}_{ij}^p = \dot{\lambda} \frac{\partial f}{\partial S_{ij}} \quad (8)$$

associated with the yield condition

$$f(S_{ij}) = \sigma_{eq} - \sigma_A(\varepsilon_{eq}^p) = 0. \quad (9)$$

Here λ is a scalar parameter, σ_{eq} and ε_{eq}^p are the equivalent stress and accumulated equivalent plastic strain

$$\sigma_{eq} = \frac{1}{\sqrt{2}} \sqrt{(S_{11} - S_{22})^2 + (S_{22} - S_{33})^2 + (S_{33} - S_{11})^2 + 6(S_{12}^2 + S_{23}^2 + S_{31}^2)} \quad (10)$$

$$\varepsilon_{eq}^p = \frac{\sqrt{2}}{3} \int_0^t \sqrt{(\varepsilon_{11}^p - \varepsilon_{22}^p)^2 + (\varepsilon_{22}^p - \varepsilon_{33}^p)^2 + (\varepsilon_{33}^p - \varepsilon_{11}^p)^2 + 6(\varepsilon_{12}^{p2} + \varepsilon_{23}^{p2} + \varepsilon_{31}^{p2})} dt \quad (11)$$

The function $\sigma_A(\varepsilon_{eq}^p)$ prescribes isotropic strain hardening in the steel substrate or aluminium matrix. The following phenomenological function can be selected:

$$\sigma_A = \sigma_s - (\sigma_s - \sigma_0) \exp(-\varepsilon_{eq}^p / \varepsilon_r^p), \quad (12)$$

where σ_0 is the yield point and σ_s is the strength, ε_r^p is the reference value of plastic strain. Model 2 in Fig. 3 describes perfect plasticity at $\sigma_A = \sigma_0$.

For a more detailed description, σ_A can be obtained from physical consideration of dislocation dynamics. σ_A is the athermal part of the stress σ_{eq} and associated with long-range obstacles to the dislocation motion. It is independent of the strain rate

and mainly depends on the microstructure of the material: the dislocation density and substructures, grain sizes, point defects and various solute atoms. The current yield stress is proposed to be defined in the following way:

$$\sigma_A = \sigma_0 + \alpha\mu b\sqrt{N(\varepsilon_{eq}^p)} + \sum_j \alpha_{1j} P_j(\varepsilon_{eq}^p), \quad (13)$$

with the Hall-Perth dependence being introduced

$$\sigma_0 = \sigma_0^0 + kd^{-1/2} \quad (14)$$

where σ_0^0 is the yield point of the single crystal and d is the grain size.

The addend in Eq. (13) is a familiar dependence from physics of plasticity that accounts for a microscopic contribution from a forest of dislocations, where N is the dislocation density. The third summand is associated with the formation of substructures: dislocation cells, band and fragmented substructures, where α_{1i} denotes coefficients accounting for dislocation substructure contributions to strengthening, and the probability functions $P_j(\varepsilon_{eq}^p)$ are connected with volume fractions of the substructures. Based on the experimental evidence the following form of the exponential law is proposed:

$$P_j(\varepsilon_{eq}^p) = \int_0^{\varepsilon_{eq}^p} \lambda_j \exp(\eta - \exp \eta) d\varepsilon_{eq}^p. \quad (15)$$

Here $\eta = -\lambda_j(\varepsilon_{eq}^p - \varepsilon_{eqj}^p)$, ε_{eqj}^p is a parameter associated with deformation giving rise to formation of the i -th substructure, and λ_j specifies the strain range wherein the substructure exists. The volume fraction of the substructure of interest is determined via the distribution function (15) to give

$$P_j^v(\varepsilon_{eq}^p) = \exp(1 + \eta - \exp \eta). \quad (16)$$

For instance, for the only dislocation cell substructure Eq. (13) takes the form

$$\sigma_A = \sigma_0 + \alpha\mu b\sqrt{N(\varepsilon_{eq}^p)} + \alpha_c P_c(\varepsilon_{eq}^p). \quad (17)$$

Comparing Eq. (17) with the well-known experimental evidence $\sigma_A \propto d_c^{-1}$ [49], which is similar to Eq. (14), and taking into account the fact that the dislocation cell diameter d_c decreases during plastic deformation reaching the saturation point where its value does not depend on the stacking fault energy ($d_c^{sat} \approx 0.2 \mu\text{m}$ for many materials [50, 51]), the following expression can be obtained

$$d_c(\varepsilon_{eq}^p) = \frac{d_c^{sat}}{P_c(\varepsilon_{eq}^p)}. \quad (18)$$

The physically-based expression (13) includes microscopic parameters such as elastic moduli, dislocation density, grain size and dislocation cell diameter, module of Burgers vector b , empirical strength coefficient α , while purely phenomenological Eq. (12) operates with macroscopic yield point and strength.

4.2 Strain Rate and Temperature Effects

To describe strain rate and temperature sensitivity of steel or aluminum, it is necessary to develop a relaxation constitutive equation (model 4 in Fig. 3). Substituting (8) in the equation for the equivalent plastic strain (11), it can be found

$$\dot{\varepsilon}_{ij}^p = \frac{3}{2} \frac{\dot{\varepsilon}_{eq}^p}{\sigma_{eq}} S_{ij}. \quad (19)$$

In order to describe the equivalent plastic strain rate $\dot{\varepsilon}_{eq}^p$, let us continue proceeding from a dislocation concept of plastic flow. Kinetic equations for the plastic strain rate based on the motion of dislocations are the subject of a considerable amount of literature. Theoretical concepts relevant to the discussion are in part summarized in [52]. Following the model let us define

$$\dot{\varepsilon}_{eq}^p = \dot{\varepsilon}_r^p \exp \left\{ -\frac{G_0}{kT} \left[1 - \left(\frac{\sigma_{vis}}{\tilde{\sigma}} \right)^w \right]^z \right\}, \quad (20)$$

$$\sigma_{vis} = \sigma_{eq} - \sigma_A(\varepsilon_{eq}^p) \quad T = T_0 + \int_0^{\varepsilon_{eq}^p} \frac{\beta}{\rho C_v} \sigma_{eq} d\varepsilon_{eq}^p.$$

Here G_0 is the energy that a dislocation must have to overcome its short-range barrier solely by its thermal activation, $\tilde{\sigma}$ is the stress above which the barrier is crossed by a dislocation without any assistance from thermal activation, k is the Boltzmann constant, $\dot{\varepsilon}_r^p$ is the reference value of the plastic strain rate, T_0 is the test temperature, C_v is the heat capacity, β is the fraction of plastic work which is converted into heat. $\beta \cong 1$, $z = 2/3$ and $w = 2$ for many metals [52].

Parameters of the model were derived by solving an initial value problem and by fitting the calculation results to the experimental stress-strain curves under tension at different strain rates and temperatures. For uniaxial loading in the X-direction $\sigma_{xx} = \sigma_{eq}$ and $\varepsilon_{xx} = \varepsilon_{eq}$. In this case the constitutive Eq. (4) takes the form

$$\sigma_{eq} = E(\varepsilon_{eq} - \varepsilon_{eq}^p), \quad (21)$$

where E is the Young's module. Equations (20), (21) and (12) were solved numerically by a fourth-order Runge–Kutta method.

The results obtained are presented in Fig. 5. In order to validate the model, Eqs. (19) and (20) were introduced into the commercial software ABAQUS and the tension of steel H418 plates was simulated in a plane stress formulation. Here and in what follows, stress $\langle \sigma \rangle$ was computed as the equivalent stress σ_{eq} averaged over the mesovolume $\langle \sigma \rangle = \sum_{k=1,n} \sigma_{eq}^k s^k / \sum_{k=1,N} s^k$, where n is the number of computational mesh cells and s^k is the k -th cell area. Strain ε corresponds to the relative elongation of the computational domain in the X-direction $\varepsilon = (L - L_0) / L_0$, where L_0 and L are the initial and current lengths of the computational domain along X. The results show good agreement between the calculations and experiment (Fig. 5d). Model parameters are presented in Table 1. Strain rate effects were not taken into account for Al6061 alloy.

For the investigated steels, $G_0/k = 10.6 \times 10^{-5} \text{ K}^{-1}$ and $\tilde{\sigma} = 1450 \text{ MPa}$ are the constants which reflect the temperature sensitivity of the material. They were extracted from experimental mechanical tests at different test temperatures. Since corresponding experiments for the austenitic steels were not available, the values of these parameters are suggested to be the same as for HSLA-65 steel defined in [52].

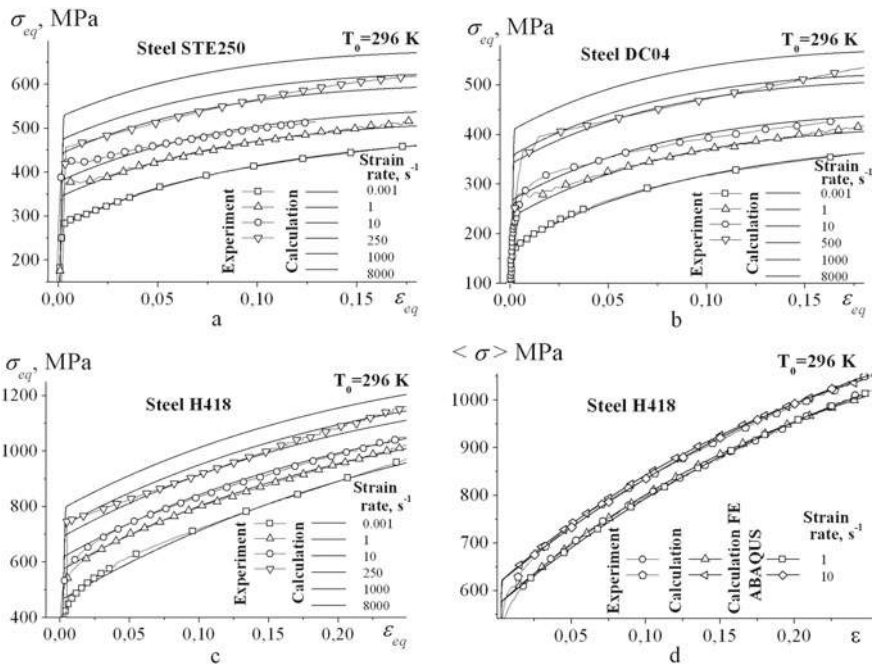


Fig. 5 Predicted mechanical properties of austenitic steels in comparison with the experimental data (a–c) and comparison of the calculation results for H418 steel with those obtained by ABAQUS (d)

Table 1 Material constants and model parameters for different steels and an Al6061 alloy

	σ_s , MPa	σ_0 , MPa	ε_r^p	$\dot{\varepsilon}_r^p$, s ⁻¹
HSLA	713	422	0.21842	4·108
H418	1300	466	0.27456	2·109
STE250	497	278	0.09978	3·1010
DC04	395	173.7	0.09312	5·1010
Al6061	184	62	0.054	–

Experimental and calculated stress-strain curves for HSLA-65 steel are presented in Fig. 6. It can be seen that for $\varepsilon < 10 \div 20\%$ Eq. (20) overestimates the current stress and fails to give a correct description of the shape of the stress-strain curve. The overestimation may be due to the fact that the parameter $\dot{\varepsilon}_r^p$ is assumed as the constant value which is proportional to the density of mobile dislocations $N_m \cong 10^{11} \text{ cm}^{-2}$ [52], although it is well known that N_m changes with the plastic deformation development and reaches its saturation at total strains of $10 \div 40\%$ for different steels.

In order to take into account the evolution of the dislocation continuum, the plastic strain rate is proposed to be as follows

$$\dot{\varepsilon}_{eq}^p = \dot{\varepsilon}_r^p F(\varepsilon_{eq}^p) \exp \left\{ -\frac{G_0}{kT} \left[1 - \left(\frac{\sigma_{vis}}{\tilde{\sigma}} \right)^w \right]^z \right\} \tag{22}$$

where

$$F(\varepsilon_{eq}^p) = F^* + (1 - F^*) \cdot \exp \left(-\frac{B}{|g|b} \varepsilon_{eq}^p \right) \tag{23}$$

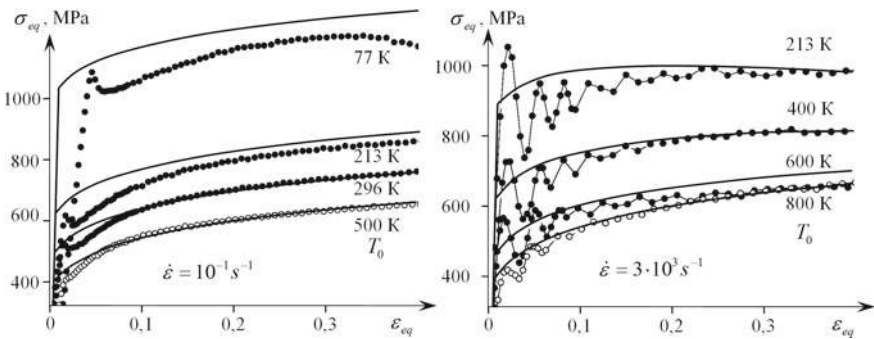


Fig. 6 Stress-strain curves for steel HSLA-65 [52]. Dots—experiment, lines—calculations, using Eq. (20)

is the fraction of mobile dislocations (Kelly and Gillis 1974) which decreases at the initial stages of plastic flow due to stalemating events. Here $|g| = 0.5$ is the orientation factor, $b \cong 3.3\text{\AA}$ is the magnitude of the Burgers vector, F^* is the minimum value of F .

It is suggested for F^* and B to be connected with the reciprocal mean free path for stalemating events as follows. Parameter B is attributed to the initial state of the material, when the free path is a half of an average grain size d

$$B = \frac{2}{dN^0}, \quad (24)$$

where N^0 is the initial dislocation density.

In the equation for F^* , the free part is calculated from some intermediate state, using the dislocation cell diameter d_c^{sat} formed inside the grains

$$F^* = \frac{N^0 d}{N^* d_c^{sat}}, \quad (25)$$

where N^* is the maximum dislocation density. In this formulation the density of mobile dislocations

$$N_m^* = N^* F^* \quad (26)$$

is a measure for the saturation density. Defining $N^* = 5 \times 10^{12} \text{ cm}^{-2}$, we obtain from Eq. (26) $F^* = 0.02$. The grain size $d = 15 \mu\text{m}$ for HSLA-65 steel [52], hence $N^0 \cong 1.33 \times 10^9 \text{ cm}^{-2}$ from Eq. (25) and $B = 0.01 \mu\text{m}$ from Eq. (24).

The system of Eqs. (1)–(7), (12) and (22) for a rectangular homogeneous region was solved numerically by the finite-difference method (Sect. 6). Figure 7 shows the results of plane strain calculations for varying strain rate and temperature. For reference, a dotted curve (296 K, 8000/s) is plotted in the figure to present calculations according to the model, where $\dot{\varepsilon}_{eq}^p$ was computed, using Eq. (20). Relation (22) is

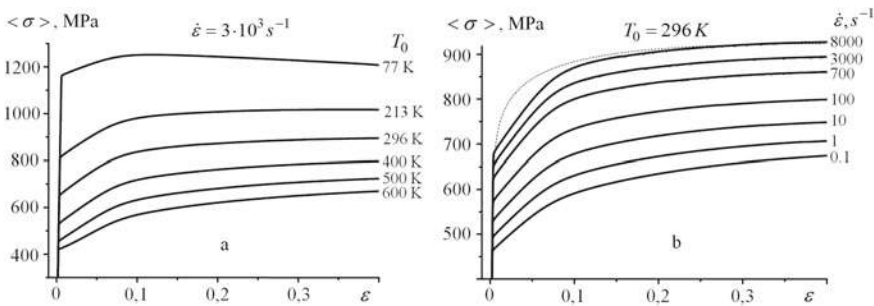


Fig. 7 Stress strain curves calculated using Eq. (22) for different temperatures (a) and strain rates (b)

seen to provide a more accurate description of stress-strain curves than Eq. (20) for small deformations ($\varepsilon \leq 20\%$).

4.3 Lüders Band Propagation and Jerky Flow

Experimental stress-strain curves for HSLA-65 steel are characterized by the upper and lower yield stresses (Fig. 6), which could be an evidence of Lüders band propagation.

The jerky-flow phenomenon in alloys has been well studied experimentally, for instance in [52–59], and attributed to the formation of localized deformation bands at the mesoscale. As a special case of such an anomalous behavior, Lüders band propagation is characterized by single displacement of macroscopic localization zone along the test piece. It is generally agreed that the microscopic essence of the discontinuous yielding is the dynamic aging of dislocations by diffusing solute atoms. There are some physically based attempts to simulate the propagation of bands of localized plastic deformation [60–62].

In this paper, to describe Lüders band propagation (model 5 in Fig. 3), use was made of a phenomenological approach [63]. It is suggested that the methods of continuum mechanics and discrete cellular automata can be used in combination. The approach relies on the experimentally established fact that the plastic deformation originates near surfaces and interfaces and subsequently propagates from the surface sources as localized deformation front.

Each cell of the computational grid is treated as a cellular automaton which can be either in elastic or plastic state. Initially all cells are elastic. Elastic-to-plastic transition of a certain computational cell is controlled by both the stress value in this local point and the deformation behavior of the neighboring computational cells. Noteworthy, different yield criteria are formulated for the surface cells and for local regions in the bulk of the material. In the former case, a local region near the surface becomes plastic if the equivalent stress acting there reaches its critical value; the stress-based criterion is used for the surface and interface cells. The response of an internal region is elastic until two conditions are satisfied: the equivalent stress in this cell achieves the yield limit and the plastic deformation accumulated in any of the neighboring cells amounts to its critical value. In such a way, the internal regions are successively involved in plastic deformation by flows propagating from the surface and interface sources.

Mathematically, the stress-based yield criterion given by a physically-based constitutive model for any local internal region D is complemented with the necessary condition whereby there must be plastic deformation at least in one of the regions D^* adjacent to D :

$$\varepsilon_{eq}^{p*} = \varepsilon_0. \quad (27)$$

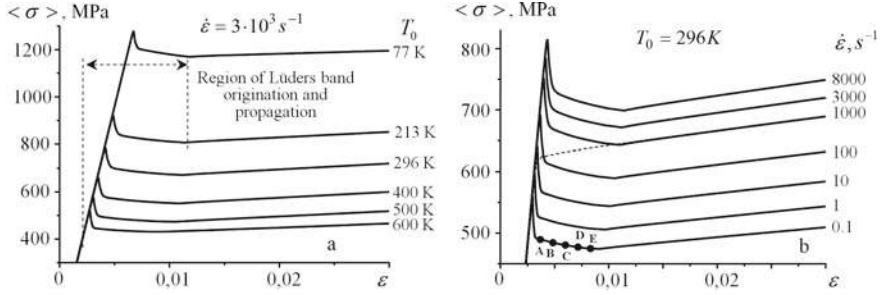


Fig. 8 Initial portions of stress-strain curves calculated using Eqs. (27) and (22) for different temperatures (a) and strain rates (b)

Here, ε_0 is the new parameter that reflects material properties associated with the strain aging effects. This is a critical value of the equivalent plastic strain accumulated in the D^* region, which is necessary for the onset of plastic flow in the region D .

Using the criterion (27) and the constitutive model (22) in combination, we have performed numerical simulations of Lüders band propagation in a wide range of strain rates and temperatures. Figure 8 demonstrates calculated stress-strain curves at early deformation stages ($\varepsilon \leq 3\%$). The computational domain is approximated by a regular mesh consisting of 600 cells, $n_x = 80$, $n_y = 20$. The parameter $\varepsilon_0 = 8 \times 10^{-4}$. For comparison, the dashed curve in Fig. 8b ($T_0 = 296$ K, the strain rate is 1000 s^{-1}) shows the initial portion of the curve calculated in the assumption of homogeneous deformation (see Fig. 7b).

The combined approach is seen to provide a more accurate description of the experimental stress-strain relations for HSLA-65 steel (see Fig. 6). Plastic flow first originates near the left boundary of the computational domain where load is applied and propagates along the specimen in the form of a localized plastic deformation front (Fig. 9a). The material ahead of the front is elastic and the plastic strain is accumulated behind the moving Lüders band front. Similar behavior was observed in experiments. Presented in Fig. 9b are the experimental data on Lüders band propagation in a steel plate surface-hardened by the electron-beam-induced deposition [64]. This process leads to the occurrence of the upper and lower yield points in the macroscopic stress-strain curve and a zone of slow variation in the current resistance to load—the yield plateau (Fig. 8).

On further loading, the stress relaxation in the elastic region slows down (Fig. 10). Simultaneously, the strain hardening in the expanding plastic flow region makes an increasingly greater contribution to the macroscopic stress. Therefore, the yield plateau appears which is characterized by a slow variation in the current resistance to deformation (Fig. 8). In this stage, the relative elongation of the specimen occurs primarily by plastic deformation of the zone located behind the front.

The approach discussed was further developed to account for the Portevin-Le Chatellier effect (model 6 in Fig. 3) associated with sequential propagation of multiple

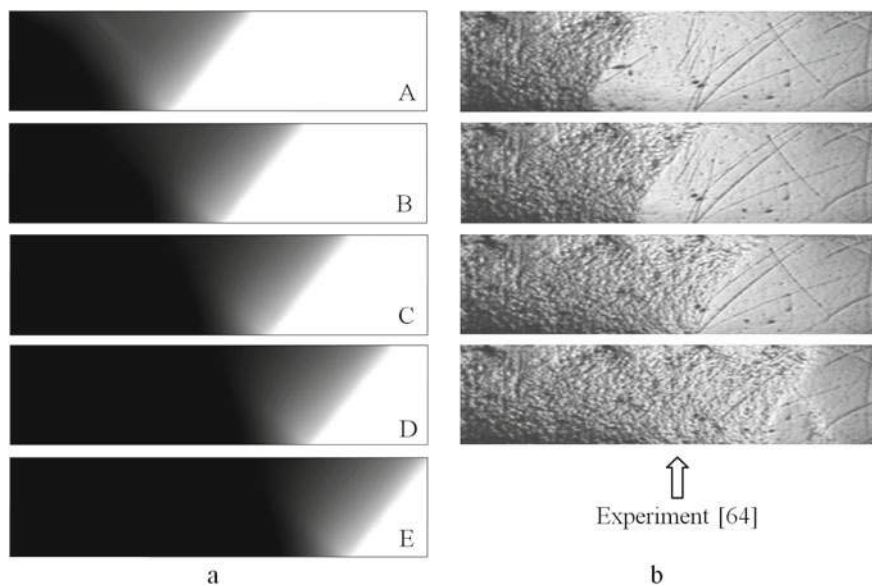


Fig. 9 Equivalent plastic strain distributions for Lüders band propagation: calculation results (a) where A–E are the material states corresponding to the points A–E in Fig. 8b and experimental data [64] (b)

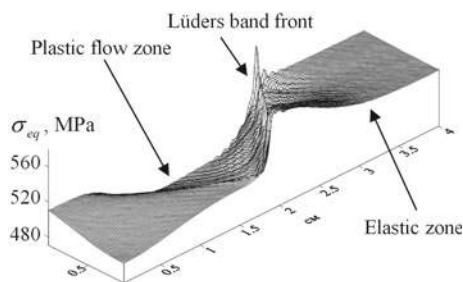


Fig. 10 Equivalent stress filed at a compressive strain corresponding to the point A in Fig. 8b

localized deformation bands from the specimen ends (model 6 in Fig. 3). Experimental observations of unstable deformation effects [55–59] show that propagation of a localized deformation band corresponds to each drop in stress seen in the stress-strain curve. As a rule, the stress amplitude at that point and the average quasi-homogeneous deformation during time intervals between the sequential band formation are found to increase with strain hardening. Therefore, condition (27) was modified to

$$\varepsilon_{eq}^{p*} = \varepsilon^*(\xi, \varepsilon_0), \quad \xi = \sigma_A(\varepsilon_{eq}, \sigma_0) / \sigma_0. \quad (28)$$

The localized-deformation bands are formed near specimen loaded ends at regular intervals with the proviso that $\Delta \varepsilon_{eq}^{p\min} = \varepsilon^\Delta(\xi, \varepsilon_0)$. Here, $\Delta \varepsilon_{eq}^{p\min}$ is a minimum increase in the equivalent plastic deformation as the result of propagation of the previous band. Thus, in the model proposed, the drop in stress and the periodic generation pattern of the localized deformation bands are expressed by some dimensionless parameter ξ accounting for the strain hardening. The simple relations

$$\varepsilon^* = \varepsilon_0 \exp\left(\frac{\xi}{1 - \xi}\right), \quad \varepsilon^\Delta = \varepsilon_0(\xi - 1), \quad (29)$$

were derived in numerical simulations of loading of an Al6061 alloy that exhibits unstable plastic flow [55]. The parameters for a hardening function of the type given in Eq. (12) were chosen according to the experiments performed in [55] (see Table 1). The calculated results demonstrated in Figs. 11 and 12 are an evidence for an essentially nonhomogeneous stressed-strained state at the yield point. The stress-strain curve is a serrated line (Fig. 11).

Each drop in the curve (Fig. 11d) corresponds to the formation and propagation of one or two localized deformation bands (Fig. 12). According to Eq. (29), the quantities $\varepsilon^*(\xi, \varepsilon_0)$ and $\varepsilon(\xi, \varepsilon_0)$ are rather small in early plastic flow stages, as a result of which the stress amplitude during the drastic decrease in the equivalent stress is

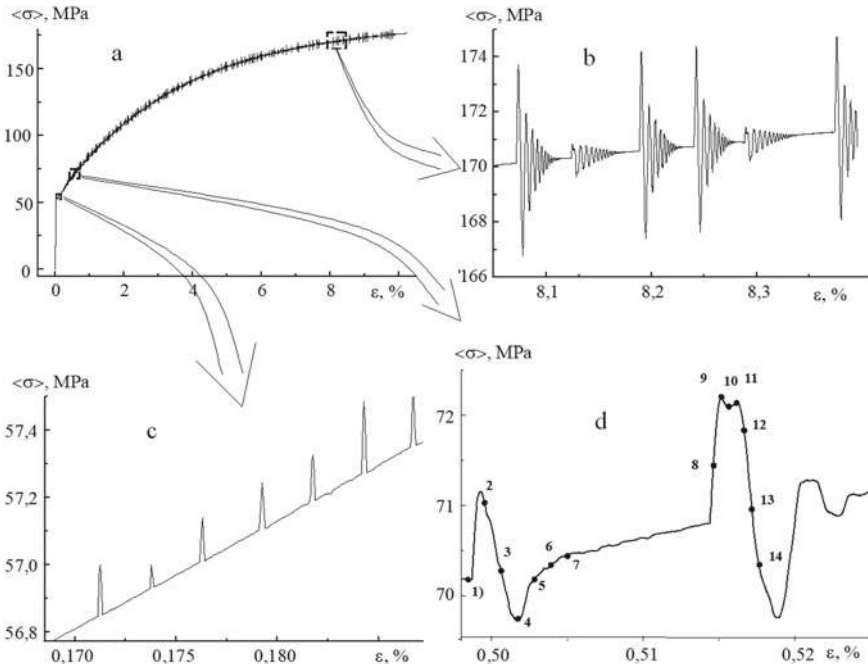


Fig. 11 Calculated stress-strain curve for an Al6061 alloy (a) and three selected sections shown at enlarged scales (b–c)

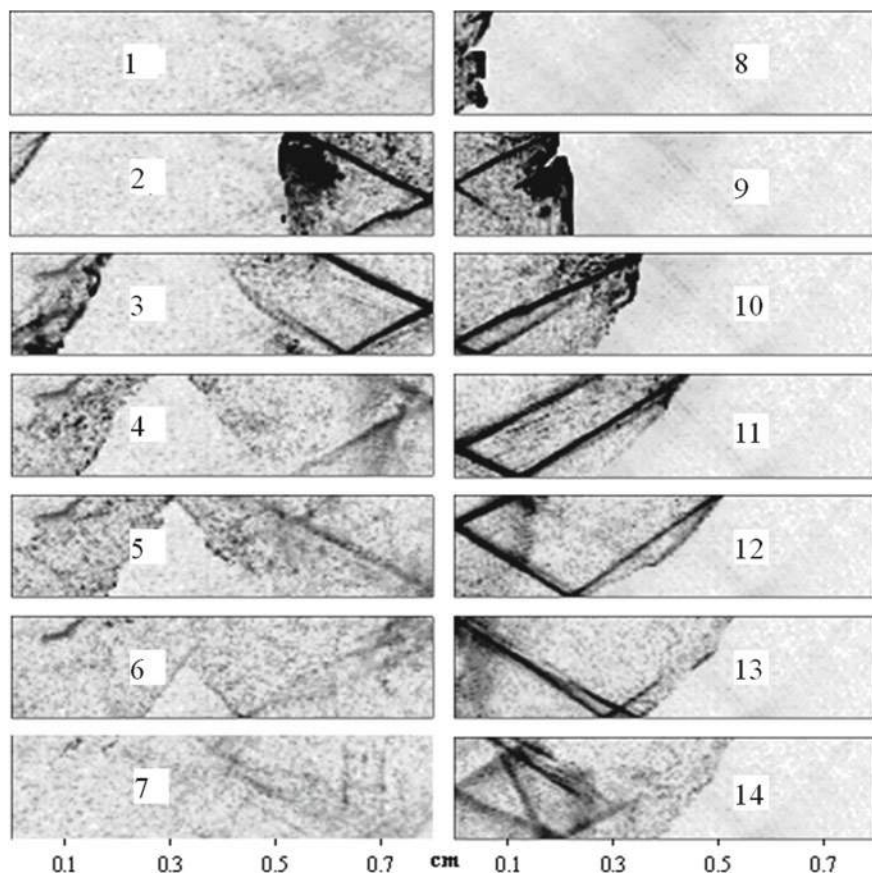


Fig. 12 Equivalent plastic strain rate distributions for strain states 1–14 in Fig. 11

low (Fig. 11c). On further loading, the deformation fronts are set in a regular motion as a consequence of hardening and nonhomogeneous deformation due to propagation of the previous localized-deformation band. The band can move faster or slower or can even cease to move (Figs. 12 and 14), which causes oscillations in the stress-strain curves (Fig. 11b and d). Propagation of one deformation band is responsible for oscillations of larger amplitude (Figs. 12 (8–14), 11d), whereas formation of two deformation fronts propagating from the boundaries of the computational domain in opposite directions is associated with oscillations of smaller amplitude (Figs. 12 (1–7) and 11d).

4.4 Brittle Fracture of Ceramic Particles and Coatings

A distinctive feature of deformation of brittle materials is the fact that under compression their fracture can occur along planes where the macroscopic stresses are thought to be zero [65], so that the crack can propagate along the direction of the applied loading (the X-direction in Fig. 13). For instance, for particle-reinforced metal matrix composites and coated materials it was experimentally shown that cracks in the particles and in the coating under compression are largely oriented along the direction of compression [66, 67]. For the case of modeling of a homogenous specimen, however, the stress tensor components in the transverse direction, Y , which are supposed to open this crack, are identically equal to zero. Thus, in experiments, cracks propagate under stresses that are zero from the standpoint of mechanics. To describe fracture in this case use is made of strain-based criteria. The simplest one is the criterion of positive elongation along the planes normal to the above-mentioned cross-section, [65], i.e., along Y .

In what follows, we are going to show that in simulation of composites the stress tensor components along Y are nonzero in contrast to the case of a homogeneous material. Moreover, at the interfaces there are localized regions oriented with respect to the direction of externally applied compression so that they experience tensile stresses. It is these stresses that can give rise to crack opening and propagation along the direction of external loading.

To describe fracture of the boride coating and corundum particles (model 7 in Fig. 3) use was made of the maximum distortion energy criterion. The criterion is thought to poorly describe fracture of brittle materials. In this work, we show that when it is applied to composite materials with realistically simulated interface geometry, where the calculations contain localized regions of tensile stresses under any type of external loading, the maximum distortion energy criterion works fairly well and provides a correct description of brittle materials and composites. We have modified the criterion to account for the difference in strength values of the tensile and compressive regions:

$$\sigma_{eq} = \begin{cases} C_{ten}, & \text{if } \varepsilon_{kk} > 0 \\ C_{com}, & \text{if } \varepsilon_{kk} < 0, \end{cases} \quad (30)$$

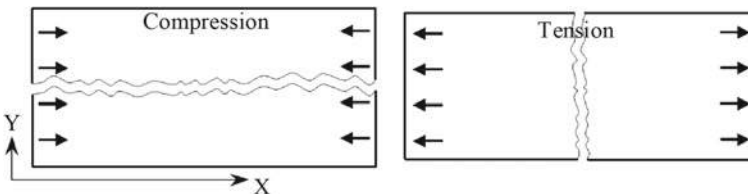


Fig. 13 Fracture under tension and compression

Table 2 Elastic modules, densities and strength constants [15, 68]

	ρ , kg/m ³	K , GPa	μ , GPa	C_{ten} , GPa	C_{com} , GPa
Steels	7900	133	80	–	–
Al6061	2700	76	26	–	–
<i>FeB</i>	7130	200	140	1	4
Al ₂ O ₃	3990	438	141	0.26	4

where C_{com} and C_{ten} are the values of the tensile and compressive strengths. According to the criterion, Eq. (30), fracture occurs in the local regions undergoing bulk tension. The following fracture conditions are prescribed for any local region of the coating: if the cubic strain ε_{kk} has a negative value and σ_{eq} reaches its critical value C_{com} , then all components of the deviatoric stress tensor in this region are taken to be zero, and in the case of $\varepsilon_{kk} > 0$ and $\sigma_{eq} \geq C_{ten}$, pressure P is equal to zero as well. Elastic modules for the composite constituents and strength parameters for the boride coating and corundum particles are presented in Table 2.

5 Finite-Difference Numerical Procedure

The boundary-value problems in terms of the plane strain were solved numerically by the finite difference method (FDM) [69, 70]. In contrast to the finite-element method (FEM), where the solution of the system is approximated, the FDM approximates the derivatives entering this system.

Let us look at a microstructure region near the base of one of the teeth (Fig. 14a). The region is approximated by a mesh containing N uniform rectangular cells $N = N_x \times N_y$ (Fig. 14b). The mesh is “frozen” into the material and is deformed together with it. The system of equations for this mesh is replaced by a difference analog. Use is made of an explicit conditionally stable scheme of the second order of accuracy. For the time step, it is necessary that the Courant criterion is satisfied as follows:

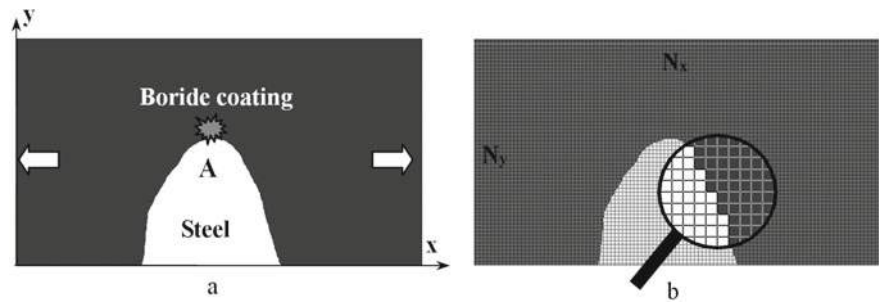


Fig. 14 A-type region of the composite structure (a) and discretization of the region (b)

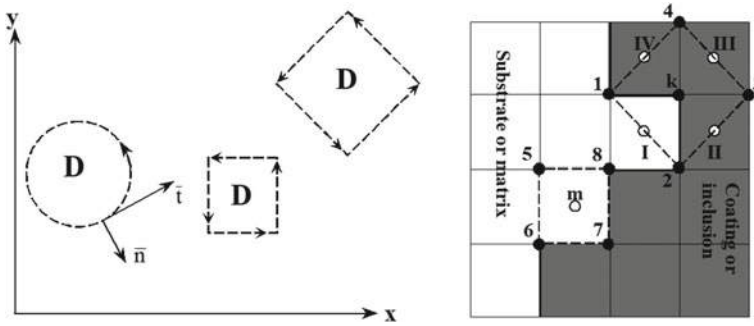


Fig. 15 Schematic representation of approximating the space derivatives

$$\Delta t = k_C \frac{h_{\min}}{C_l}, \quad (31)$$

where h_{\min} is the minimum step of the mesh, C_l is the longitudinal velocity of sound, and $0 < k_C < 1$ is the Courant ratio. The stability condition, Eq. (31), implies that an elastic wave within one time step does not cover the distance longer than the minimum mesh step.

The values of stress σ_{ij} , strain ε_{ij} , and density ρ are computed in the cell centers (points I, II in Fig. 15), while those of the displacements u_i and velocities \dot{u}_i correspond to the mesh nodes (points 1, 2 in Fig. 15). Let us use the following definition of partial derivatives:

$$\frac{\partial F}{\partial x} = \lim_{D \rightarrow 0} \frac{\int_B F(\bar{n} \cdot \bar{i}) ds}{D}, \quad \frac{\partial F}{\partial y} = \lim_{D \rightarrow 0} \frac{\int_B F(\bar{n} \cdot \bar{j}) ds}{D}, \quad (32)$$

where B is the boundary of region D , s is the arc length, \bar{n} is the normal vector, and \bar{i} is the tangent vector (Fig. 15).

$$\bar{n} = \frac{\partial x}{\partial n} \bar{i} + \frac{\partial y}{\partial n} \bar{j} = \frac{\partial y}{\partial S} \bar{i} - \frac{\partial x}{\partial S} \bar{j}. \quad (33)$$

Applying Eqs. (32)–(33) to the rectangular regions bounded by dashed lines (Fig. 15), we obtain for the case of stress derivatives corresponding to, e.g., node k

$$\int_C \sigma_{ij}(\bar{n} \cdot \bar{i}) ds = \int_C \sigma_{ij} \frac{\partial y}{\partial S} ds,$$

from which

$$(\sigma_{ij,x})^k = \frac{1}{D} (\sigma_{ij}^I (y_2 - y_1) + \sigma_{ij}^{II} (y_3 - y_2) + \sigma_{ij}^{III} (y_4 - y_3) + \sigma_{ij}^{IV} (y_1 - y_4)) \quad (34)$$

and

$$\int_C \sigma_{ij} (\bar{n} \cdot \bar{j}) ds = \int_C \sigma_{ij} \frac{\partial x}{\partial S} ds,$$

whence

$$(\sigma_{ij,y})^k = \frac{1}{D} (\sigma_{ij}^I (x_2 - x_1) + \sigma_{ij}^{II} (x_3 - x_2) + \sigma_{ij}^{III} (x_4 - x_3) + \sigma_{ij}^{IV} (x_1 - x_4)). \quad (35)$$

Derivatives $u_{i,j}$, in their turn, correspond to the cell centers and are calculated from the values of u_i in the surrounding nodes. In particular, for cell m , we have

$$(u_{i,x})^m = \frac{1}{2D} \begin{pmatrix} (u_i^5 + u_i^6)(y_6 - y_5)(u_1^6 + u_i^7)(y_7 - y_6) + \\ (u_i^7 + u_i^8)(y_8 - y_7)(u^8 + u_i^5)(y_5 - y_8) \end{pmatrix} \quad (36)$$

$$(u_{i,y})^m = \frac{1}{2D} \begin{pmatrix} (u_i^5 + u_i^6)(x_6 - x_5)(u_1^6 + u_i^7)(x_7 - x_6) + \\ (u_i^7 + u_i^8)(x_8 - x_7)(u^8 + u_i^5)(x_5 - x_8) \end{pmatrix} \quad (37)$$

where D is the area of the respective quadrangle. The computation is performed in time steps, moving from one layer n to another $n + 1$.

$$\dot{u}_i = \frac{u_i^{n+1} - u_i^n}{\Delta t} \quad (38)$$

In modeling multi-phase materials, the interface between the microstructure constituents goes across the computational mesh nodes (Fig. 15, thick solid line), with the properties of the two materials prescribed on either side of this interface. The continuity of displacements and normal stresses at this interface is, therefore, preserved, i.e., the conditions of an ideal mechanical contact are satisfied. It is shown in [69] that in the case where the densities of the two materials differ only slightly, the approximating equations, Eqs. (34) and (35), at this interface could be used without any changes. Inaccuracies might appear if we consider, e.g., a solid–liquid interface. In the latter case, it is reasonable to use a formula taking into account both left and right limits in the approximation.

6 Coated Materials

6.1 Overall Plastic Strain and Fracture Behavior Under Tension of the Coated Material with Serrated Interface

In this Section a model microstructure of coated steel DC04 (Fig. 5) with a toothed interface is investigated under tension (Fig. 16a). Figure 16b illustrates a macroscopic stress-strain curve for the mesovolume. The stages in the stress-strain curve are due to the fact that the substrate and coating materials behave in different ways in the corresponding deformation stages: 1—the substrate and coating experience elastic strain, 2—plastic flow develops in the substrate, whereas the coating is still in the elastic state, 3—a cracking of the coating.

Because of the difference in the elastic moduli between the coating and the substrate, the stress and strain distributions are nonuniform even at stage 1. At stage 2, in turn, two stages of plastic deformation can be distinguished. Initially, plastic strains are initiated near the stress concentrators at the root of boride teeth (Fig. 17a). As the loading increases, plastic deformation propagates deep into the specimen, gradually covering the steel-base regions between the boride teeth (Fig. 17b). Thus,

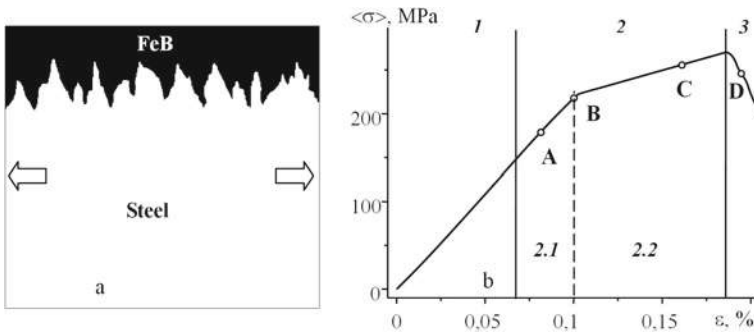


Fig. 16 Computational mesovolume (a) and its stress-strain curve under tension (b)

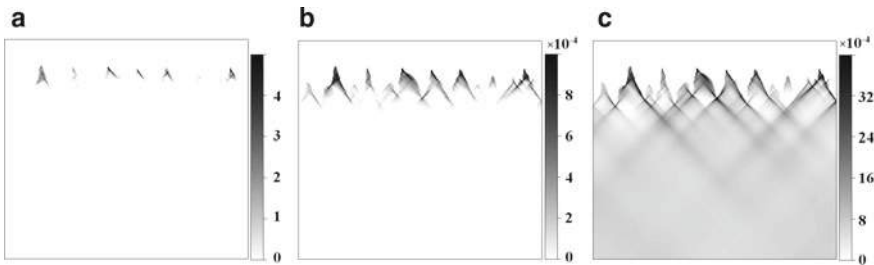


Fig. 17 Equivalent plastic strain distributions for tensile strains of 0.08 (a), 0.1, (b) and 0.16% (c) shown in Fig. 16

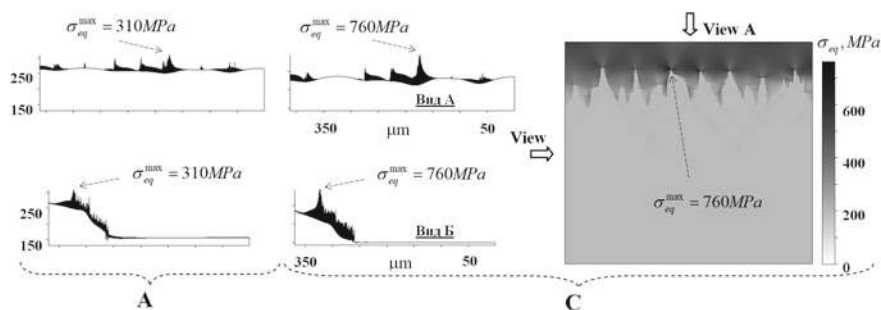


Fig. 18 Equivalent stress distributions at tensile strains lettered by A and A in Figs. 16 and 17

at stage 2.1 (Fig. 16b), plastic flow is localized near the teeth-concentrators and the main part of the base material is still in the elastic state. When an average stress level in the steel base exceeds the yield point, the main part of the base material becomes plastic—stage 2.2 is realized (Fig. 16b) at which the macroscopic stress-strain curve sharply changes the slope. As the plastic flow develops from stage 2.1 to stage 2.2, localized shear bands are formed from the stress concentrators near the boride teeth (Fig. 17c). The bands develop at an angle of about 45° to the axis of loading. Under further loading the average level of plastic strain in the substrate, as well as the degree of strain localization in the bands is increased.

A similar conclusion can be drawn relative to a value of the stress concentration at the “coating–base material” interface. Local concentrations of stresses arise at stage 1. Their distribution is due to the geometry of boride teeth. At the elastic stage, values of local stresses relative to an average level of the equivalent stress do not change in the coating. As plastic flow in the base material develops, the stress concentration increases. At stage 2.1, this effect is less pronounced—the patterns given in Fig. 18a are close to the corresponding distributions at stage 1. An intensive straining of the base material at stage 2.2 leads to a sharp nonlinear increase of local stresses (Fig. 18c).

In this case, the rate of increase of the equivalent stresses can differ for various concentrators. For example, we can see from Fig. 18 that at the stage close to elasticity three stress concentrators lying at the center of the investigated region have almost the same power (Fig. 18a). As plastic flow develops, the stresses in one of these concentrators increase relatively quicker and at the prefracture stage reach a maximum (Fig. 18c).

Stage 3 is the stage of composite failure. Plastic deformation in the base material and cracking in the coating develop simultaneously. These processes are interrelated and interconsistent. When σ_{eq}^{max} exceeds a value of C_{ten} , a first local fracture zone forms in the coating. The surrounding regions of the material begin to intensely deform and the crack propagates toward the free surface perpendicular to the direction of tension (Fig. 19). The crack formation is a dynamic process: there appears a new free surface from which release waves propagate, causing an unloading of the coating material (Fig. 19a). A descending portion is observed on the stress-strain

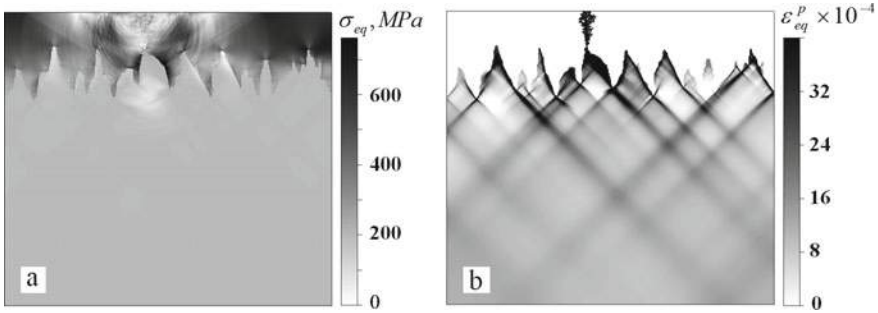


Fig. 19 Equivalent stress (a) and plastic strain distributions (b) (cracked regions in the coating are given in black) for the state D presented in Fig. 16. Total strain—0.19%

curve (Fig. 16, stage 3). A localized plastic flow enhances in the steel base near the place of crack insipience (Fig. 19b).

6.2 Interface Asperities at Microscale and Mesoscale I. Convergence of the Numerical Solution

Let us examine the loading of two different-scale regions of the coated H418 steel (Fig. 5). The first region is shown in Fig. 14. It is an A-type region at mesoscale I (see Fig. 4). The results of calculations under tensile conditions are given in Figs. 20 and 21. The crack nucleates at the hump of the convexity and propagates towards the specimen surface. Crack propagation is a purely dynamic process occurring at velocities approximating that of sound. The time of crack propagation from the interface to the specimen surface is small compared to the characteristic time of quasistatic loading. The cracking can be regarded as a formation of new free surfaces, from which elastic release waves begin to propagate. The wave dynamics of crack propagation is clearly shown in Fig. 20, where equivalent stress patterns are presented.

To verify the solution convergence, a set of calculations was performed with the step varying in space (Figs. 20 and 21). The total number of cells N in the computational mesh in each case was: 1222, 5035, 8820, 12,768, 19,950, 35,420 and 79,800. The computations showed that the fracture region has a physically-based size that is controlled by the interface geometry—its curvature, and only weakly depends on the size of a local fracture region.

Shown in Fig. 21a are the respective stress-strain curves. The drooping part of the curves corresponds to the initiation and propagation of a unit crack. Figure 21b depicts the dependence of the maximum value of $\langle \sigma \rangle$, corresponding to the onset of crack propagation, on the computational mesh size. It is evident that the solution is convergent and well approximated by the exponential law (dotted line)

$$\langle \sigma \rangle_{\max} = \sigma_{act} + \sigma_{mesh} \exp\left(N_x / N_{ref}\right), \quad (39)$$

where $\sigma_{act} = 556$ MPa is the actual stress, $\sigma_{mesh} = 140$ MPa is the mesh dependent overestimation and $N_{ref} = 79$ is the reference number of cells.

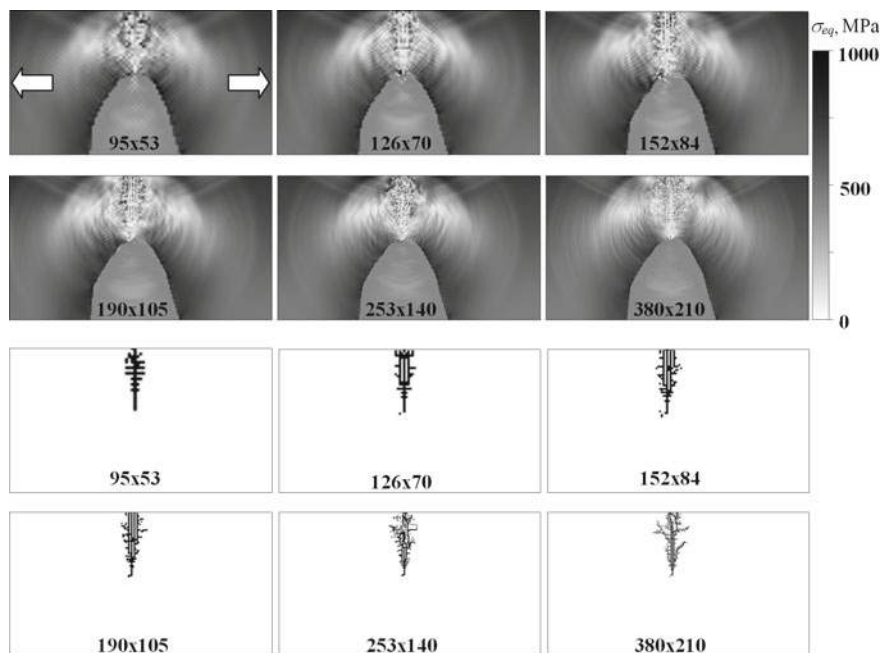


Fig. 20 Mesh-dependent equivalent stress and fracture patterns in the region presented in Fig. 14 in tension

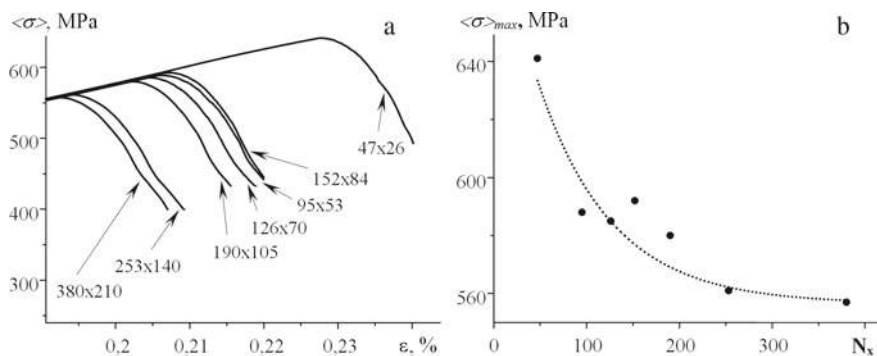


Fig. 21 Homogenized stress-strain curves (a) and the stress maximum values versus the number of cells in the X-direction (b)

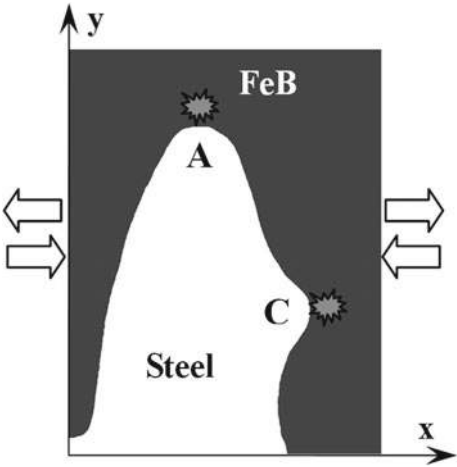


Fig. 22 Calculated region containing characteristic A- and C-type inhomogeneities

For the second computational run, we selected a larger region (Fig. 22). This region is part of the structure given in Fig. 4; it includes the region in Fig. 14, and contains both A- and C-type inhomogeneities. Figure 23 is an illustration of the computation results under different types of external loading: tension and compression. It is evident that initiation and propagation of cracks under tensile and compressive loading occurs in different places. Under tension, cracks primarily propagate from the teeth to the specimen surface, while under compression—from one lateral face of a tooth to the other. This phenomenon is discussed in Sect. 7.4 in details.

The investigations of mesh convergence showed that in the case where a step in space is quite small, in other words when an A-type characteristic convexity of the

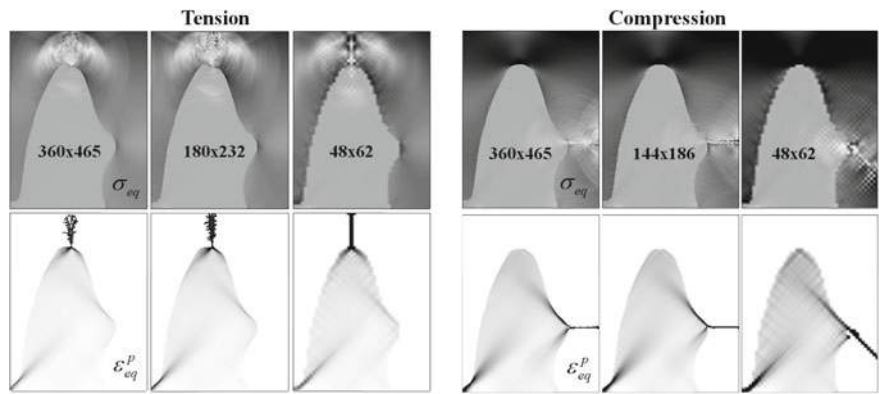


Fig. 23 Mesh-dependent equivalent stress and fracture patterns for a region containing A- and C-type inhomogeneities under tension and compression

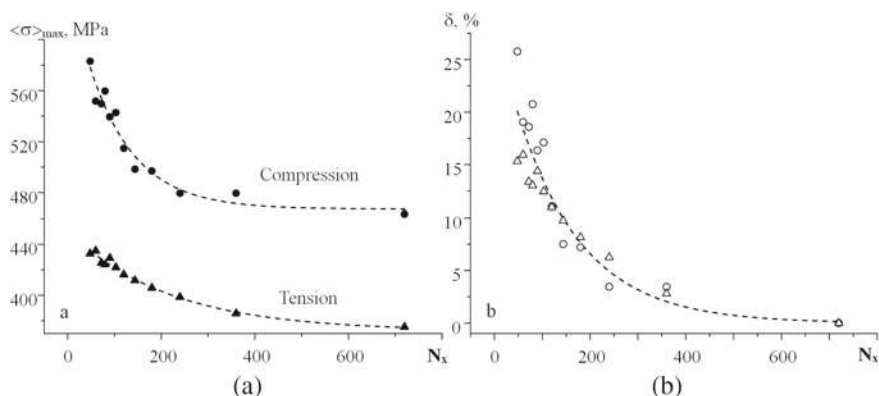


Fig. 24 Stress maximum values under tension and compression (a) and an average error (b) versus the number of cells in the X direction

Table 3 Convergence exponential parameters for the crack initiation in the coated steel

	σ_{act} , MPa	σ_{mesh} , MPa	N_{ref}
Tension	372	79	217
Compression	467	182	97

least curvature is approximated by as many as 10 computational cells, the character of fracture changes, but only slightly. In Fig. 24a the curves under both tension and compression are presented with the convergence parameters shown in Table 3. Figure 24b shows an averaged estimation of the convergence. An average error δ connected with the mesh size effects is calculated as

$$\delta = \frac{\langle \sigma \rangle_{\max} - \sigma_{act}}{\sigma_{act}} \cdot 100\%. \quad (40)$$

The dependence is approximated by the following formula (dotted line in Fig. 24b)

$$\delta = 28.5 \exp\left(-\frac{N_x}{136}\right). \quad (41)$$

The calculations discussed in this paragraph yield the following conclusions. Interface convexities are sources of unit cracks in the coating. The mesh-convergence analysis showed that the solution for this system of primary cracks converges when the step in space is decreased. The finer the mesh, the more detailed the fracture pattern, while the general behavior is similar for different meshes. There are limiting real stress and strain values for the onset of fracture, which are controlled by the physical geometry of concavities and their curvature. For the investigated microstructure and properties of contacting materials, the maximum mesh size error is about 28%.

6.3 *Fracture of the Coating with Plane Interface. Macroscale Simplification*

In this Section, the fracture criterion, Eq. (30), is shown to provide an adequate description of directions of crack propagation under different types of external loading: tension and compression. Let us simulate tension and compression of a specimen with a plane interface, i.e., excluding the interface curvature factor. Since the interface is plane, there is no concentration of tensile stresses. Fracture would not appear locally as we have excluded the cause for its local initiation. In this case, we have to artificially nucleate cracking by initially introducing a single fractured zone at the steel–boride coating interface. Even if an interface does not exhibit any serrated structure and appears to be regular at a certain scale, in the places of its adhesion to the substrate there could be various inhomogeneities, including cracks and discontinuities.

In Fig. 25, we show the results of computations under tensile and compressive loading. The differences in the direction of crack propagation are clearly seen: propagation is along the interface in the case of compression and perpendicular to it in the case of tension. For the latter, the coating separates along the interface due to the absence of boride teeth grown into the steel substrate. Thus, the fracture criterion, Eq. (30), correctly describes the direction of crack propagation in a brittle material, and its use, combined with an explicit interface of a complicated geometry, would allow us (as it will be shown later in Sect. 7.4) to interpret a possible mechanism of fracture initiation. This mechanism, according to our simulation results, is associated with the mechanical concentration of tensile stresses in the places where the interface curvature changes.

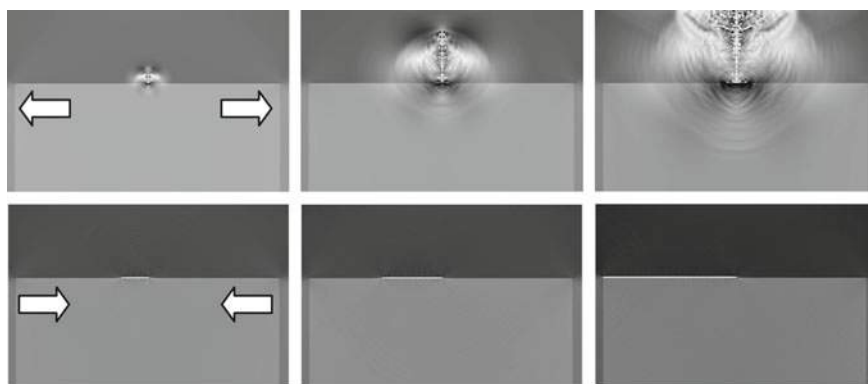


Fig. 25 Crack propagation in the coating with a plane interface under tension and compression

6.4 Plastic Strain Localization and Fracture at Mesoscale II. Effects of the Irregular Interfacial Geometry Under Tension and Compression of Composites

In the following sections we consider a real microstructure of a coated material with an interface of irregular geometry (Fig. 4b). Steel STE250 (Fig. 5) was selected as a substrate to be coated by diffusion borating. A macroscopic response of the composite microstructure under different external loading conditions is presented in Fig. 26.

According to the calculated dependences, the coated material shows higher tolerance to compressive stresses than to external tensile loading: the macroscopic yield strength, both in terms of stress and strain, is higher in the case of compression (see Fig. 26). This mechanical behavior is typical for composite materials and, as the calculations show, is associated with the principal difference of fracture processes in the coating under different external loading conditions.

Figure 27 shows the distribution of the stress tensor components under external compressive loading for the cases of needle-like and plane “coating-substrate” interfaces. It can be seen that under uniaxial loading of a coated material with the plane interface along the X-direction, only the values of σ_{xx} remain nonzero throughout the computational domain. In contrast, the serrated shape of the substrate–coating interface favors the development of a complex stressed state with nonzero values of the stress component σ_{yy} . Note that it is in the Y-direction that the material experiences both compressive and tensile stresses which are in their absolute values comparable to the values of external compressive loads. Thus, the regions of the steel substrate

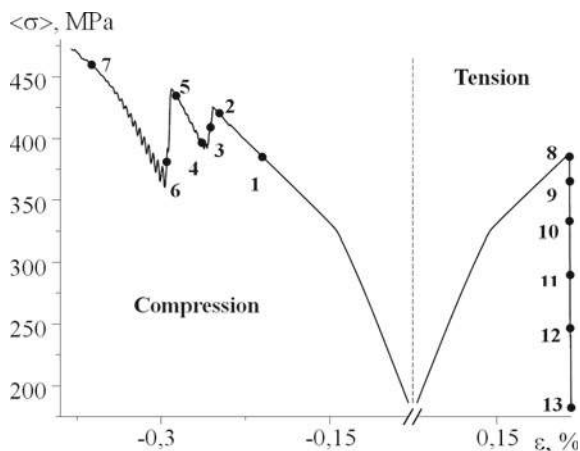


Fig. 26 Calculated stress-strain curves of the coated material under tension and compression. The respective stress and strain distributions in the mesovolume for states 1–13 are given in Figs. 27, 28 and 29

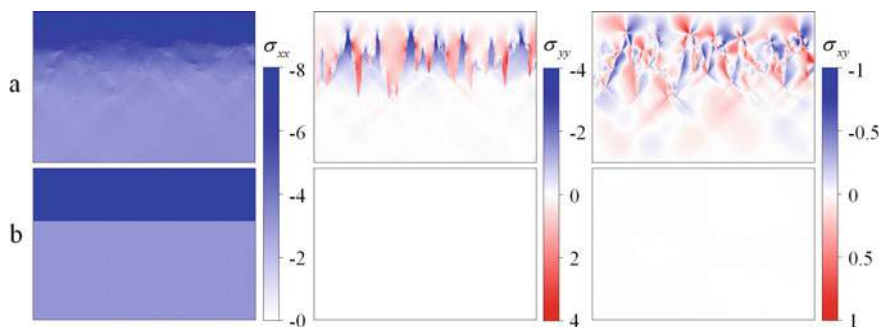


Fig. 27 Distributions of stress tensor components ($\times 100$ MPa) for needle-like (a) and plane “coating-substrate” interfaces (b) at a strain corresponding to point (1) in Fig. 26

located between the boride teeth are subjected to compressive stresses while the teeth themselves experience tensile stresses.

Should we change the direction of external loading and address tension rather than compression, the pattern presented in Fig. 27 would be the same both qualitatively and quantitatively, the difference being in the sign of the stress tensor components. Thus, the local tensile stresses develop in different places under tension and compression. This fact is responsible for the difference in fracture processes under tension and compression (Fig. 28).

Both tension and compression cracks originate in the local tensile regions. Under compression, the regions are situated at the lateral side of the boron teeth (red color regions in Fig. 27). Cracks successively nucleate on boride tooth sides and propagate along the axis of compression (Fig. 28, states 2–7). No formation of the main longitudinal crack is, however, observed. The upper coating layer maintains its stressed state and resists to the load, while multiple cracking of a boride tooth unloads the composite in the intermediate sublayer (Fig. 28, state 7). Thus, the presence of a serrated structure grown into the steel substrate prevents the coating from spalling. The stress-strain curve in this case exhibits local drops of the averaged stress, whose general level, however, continues to increase, and no catastrophic loss of strength is observed (see Fig. 26b).

A different fracture pattern is found under external tension (Fig. 28, states 8–13). The crack nucleates in the local region of highest concentration of tensile stress, which is situated at a boron tooth base, and propagates in the boride coating towards the free surface of the specimen. This unloads the material along the direction of applied tension. A descending portion appears in the stress-strain curve (points 8–13 in Fig. 26).

The formation of longitudinal and transverse cracks was found experimentally during nanoindentation (Fig. 29b), which, due to its specific geometry, gives rise both to tensile and compressive conditions in the coating within one and the same experiment. Cracks in this experiment propagate in different ways: perpendicular and parallel to the direction of applied tensile and compressive loading, respectively.

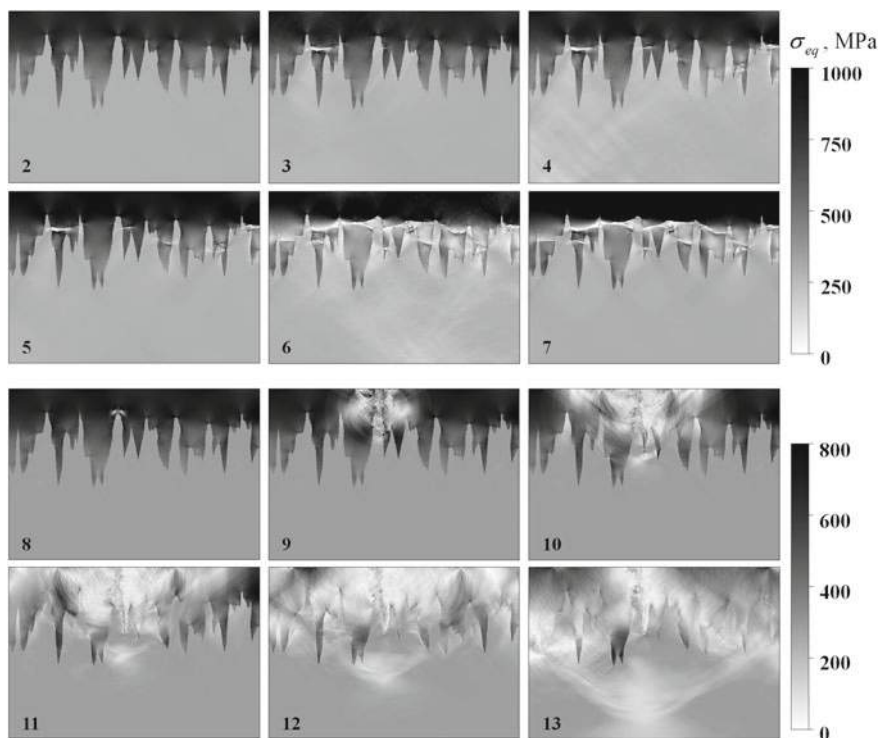


Fig. 28 Equivalent stress distributions for compressive (2–7) and tensile (8–13) strains (cf. Fig. 26)

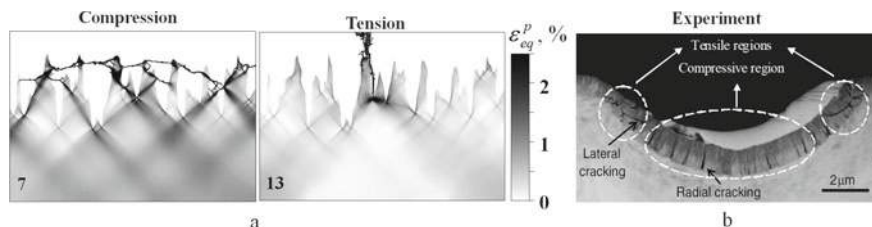


Fig. 29 Equivalent plastic strain distributions for states 2 and 13 (see Figs. 26 and 28) (a) and microscopic section of a TiN coating deposited on stainless steel after nanoindentation [67] (b)

The same result was obtained in the discussed simulations of two different experiments on uniaxial compression/tension (Fig. 29a). Note that the key role belongs to the complicated geometry of the interface and the presence of regions undergoing localized tensile stresses.

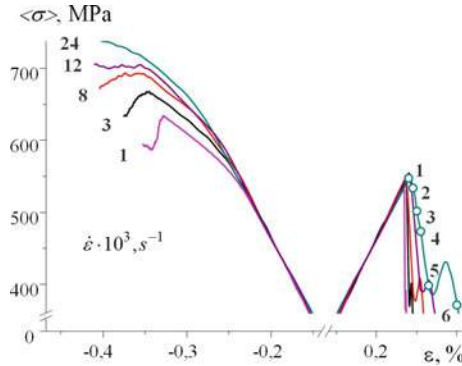


Fig. 30 Homogenized stress-strain curves under compression and tension at different strain rates

6.5 Dynamic Deformation of the Coated Material

To describe the mechanical response of the steel substrate under dynamic loading use was made of the constitutive Eq. (20) taking into account plastic strain rate and temperature in an explicit form. This equation allows a prediction of the mechanical properties of austenitic steels (Fig. 5). Boron coating is elastic-brittle. Experimental evidence shows that the elastic modules as well as the strength of brittle ceramics weakly depend on the strain rate. Therefore, strain rate sensitivity for the boron material was not introduced into the model formulation.

A series of numerical compression and tension tests were carried out by varying the strain rate externally applied (Fig. 30). The calculations show a difference between the composite responses under different types of loading: tension and compression. The following conclusions can be made.

First, under the high-strain-rate tension the fracture process intensifies in comparison with that observed under quasistatic loading. Figure 31 shows the simulation of crack initiation and growth under strain rate of $24 \times 10^3 \text{ s}^{-1}$. Comparing the results with those presented in Fig. 28 (states 8–13), it can be seen that under quasistatic loading only one crack propagates, whereas under a high strain rate multiple cracks arise. The stress under a high strain rate increases rapidly, and release waves from the first crack formation are not in time for unloading the nearby A-type stress concentration regions, i.e. the equivalent stress in one of these regions reaches the strength value C_{ten} before the release wave arriving (see Fig. 31, states 2–3).

The second conclusion is that the macroscopic strength under tension changes, but only slightly with the strain rate increasing, while both the total strain and homogenized stress of the fracture onset strongly depend on the value of the compression strain rate (Fig. 30). Figure 32 shows the stress, plastic strain and fracture patterns under compression at different strain rates. The simulations show that the higher the strain rate, the less intensive the cracking of the coating and, as a result, the higher the dynamic strength of the coated material (see Fig. 30). The explanation is the

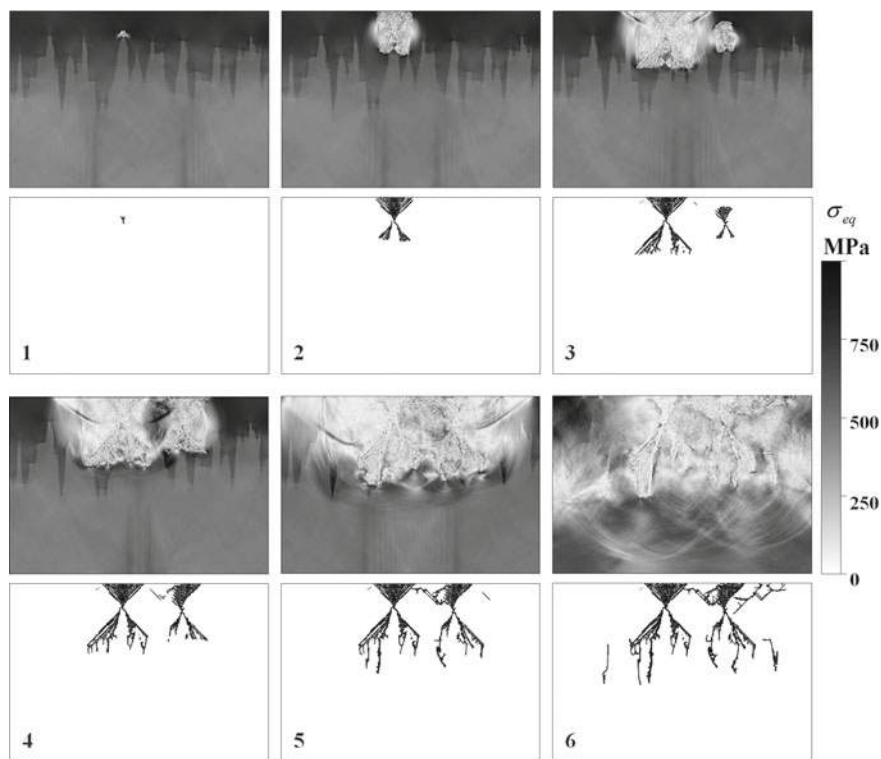


Fig. 31 Coating cracking at a strain rate of $24 \times 10^3 \text{ s}^{-1}$. Equivalent stress and fracture patterns correspond to states 1–6 shown in Fig. 30

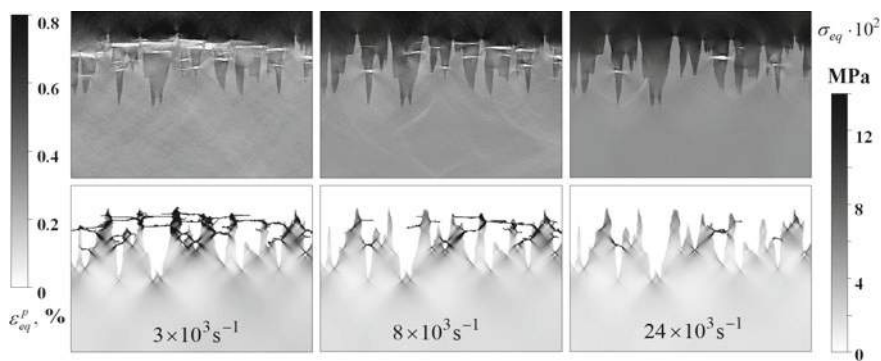


Fig. 32 Distributions of equivalent stresses and the equivalent plastic strains (fractured regions in the coating are marked by black color) for different strain rates of compression. Total strain—0.37% (see Fig. 30)

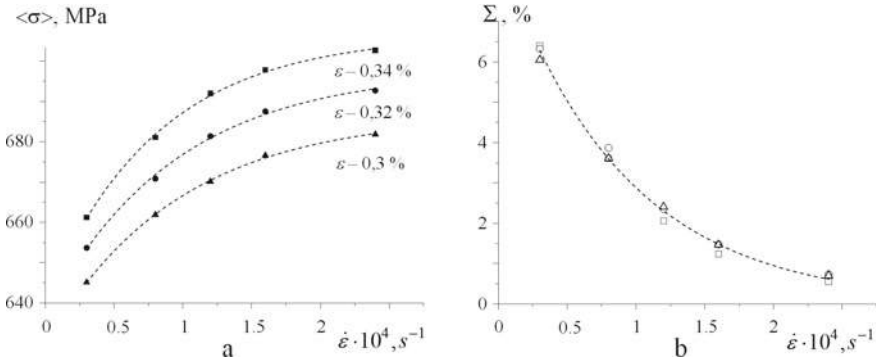


Fig. 33 Homogenized stress values for different compressive strains (a) and relative increase in the stress (b) versus the strain rate of compression

following. The value of stress concentration in C-type regions depends on the difference between mechanical properties of the steel and boron ceramics. According to the model formulation, the stress in the boron material does not change but the current dynamic yield stress of the steel increases with the strain rate increasing (Fig. 5). Therefore, the stress difference at a certain macroscopic strain decreases, and this arises only due to plasticity of the steel substrate. As the analysis of the calculation results showed, the fracture of the coating under tension develops at the elastic stage of composite deformation. The difference discussed does not change with the strain rate increasing at the elastic stage, and there is no change in the macroscopic strength.

The dependence of the macroscopic stress on the compression strain rate is shown in Fig. 33a for different strains (Fig. 30). The curves are well approximated by the exponential law (dotted lines)

$$\langle \sigma \rangle = \sigma_{dyn} - (\sigma_{dyn} - \sigma_{stat}) \exp\left(-\dot{\epsilon}/\dot{\epsilon}_{ref}\right), \quad (42)$$

where σ_{dyn} is the saturation dynamic stress, σ_{stat} is the stress under quasistatic loading, and $\dot{\epsilon}_{ref}$ is the reference strain rate. Considering the relative value, the formula can be rewritten as

$$\Sigma = \Delta \Sigma \exp\left(-\dot{\epsilon}/\dot{\epsilon}_{ref}\right), \text{ where } \Sigma = \frac{\sigma_{dyn} - \langle \sigma \rangle}{\sigma_{dyn}}, \Delta \Sigma = \frac{(\sigma_{dyn} - \sigma_{stat})}{\sigma_{dyn}}.$$

The obtained dependence presented in Fig. 33b shows that there is a possibility to predict the strain rate dependent increase in the compressive strength of a composite based on the quasistatic experiments.

7 Metal-Matrix Composites

A mesovolume of a metal matrix composite is represented schematically in Fig. 34a. The microstructure was chosen according to the experiments reported in [35].

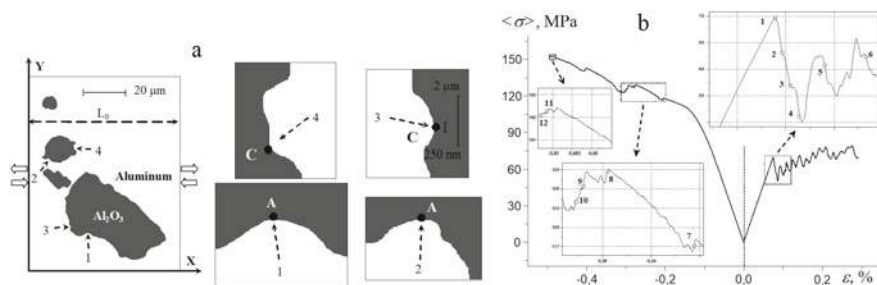


Fig. 34 Calculated composite microstructure (a) and its tensile and compressive stress-strain curves (b)

Figure 34b illustrates the macroscopic response of the mesovolume in tension and compression. Similar to the coated material, the composite under study is seen to withstand a higher load in compression than in tension. At first glance, this difference could be due to the substantial difference in the tensile and compressive strength values of the particles: since $C_{com} \gg C_{ten}$, it must take a much higher average stress level for local regions to fail in compression than in tension. However, a proportional increase in the macroscopic yield stress (relative to the ratio between aluminum and alumina volume fractions) does not occur for two reasons. First, plastic deformation in aluminum causes the total stress level to decrease (stress deviator is constrained), and free surfaces are a hindrance to a build-up of pressure. Second, the calculations for Al/ Al_2O_3 under different types of applied loading have shown that C_{com} is not reached.

Again, for the composite the microstructural inhomogeneity and interfacial effect are responsible for the formation of tensile regions, while the mesovolume is subjected to compression, and it is essential that the required values of tensile stresses are obtained in these regions. This conclusion is of critical importance for an analysis of the resulting plastic deformation of the matrix and of crack growth in brittle particles. It is necessary to stress that in simulation of uniaxial loading of a homogeneous material the stress tensor components along the direction perpendicular to the loading axis are assumed to be zero over the entire region of interest.

In the case of the metal-matrix composite the following types of inhomogeneities referred to the meso II, meso I, and microscales by their characteristic sizes can be singled out: (1) $\approx 20 \mu m$ for the particles, (2) $\approx 2 \mu m$ for the matrix-particle interfacial asperities, and (3) $\approx 250 nm$ for a local fracture zone. The particle is a mesoscale stress concentrator responsible for the formation of macroscopic localized shear bands in the matrix. The type 2 inhomogeneities will result in the stress concentration at the mesoscale I. As a consequence, plastic deformation will be localized in the matrix in the vicinity of interfacial asperities, and primary fracture zones will be formed in the particle.

Ideally the inhomogeneities can be assumed to take the form of a true circle. Hence, the stress concentration and corresponding types of the stressed state can be estimated using an analytical solution obtained in [71] for a round inclusion

embedded in a matrix. The materials for the matrix and inclusion are chosen arbitrarily. For inhomogeneities of type 1, a solution for a stiff inclusion surrounded by a comparatively soft matrix material is valid. For inhomogeneities of type 2, on the contrary, a relatively soft inclusion will be found in a rigid matrix. Finally, in the limiting case 3, we will have to solve the classical elasticity theory problem on the influence of a round hole on the stress distribution in a plate (Fig. 35). Our analytical and numerical estimations have shown that the maximum values of σ_{eq} for $\varepsilon_{kk} > 0$ are obtained at points of the A-type in tension, and at points of the C-types in compression (Fig. 35).

The first fracture zone nucleates in the vicinity of a hump of the interface concavities (Fig. 35). The fracture zone is a new stress concentrator. A new interface between fractured material and ceramics is characterized by a higher curvature due to a smaller area and by a larger difference in mechanical characteristics than the aluminum–alumina interface, since the fractured material no longer resists shear. This is a more powerful stress concentrator at the microscale than a concavity at the mesoscale I but it is formed following the same principle. Under external tension, the maximum value of the equivalent stress is observed in the A type tensile regions near the new fractured material—alumina interface (Fig. 35a), while under external compressive loading it is the C regions that undergo tensile stresses, in whose vicinity the second fracture zone is formed (Fig. 35b). Further, the process is repeated, and the crack propagates perpendicular to the direction of applied loading in the case of tension, and parallel to it—under compression.

Deformation and fracture of the composite are illustrated in Fig. 36, where the equivalent stress and strain distributions and velocity fields superimposed on a

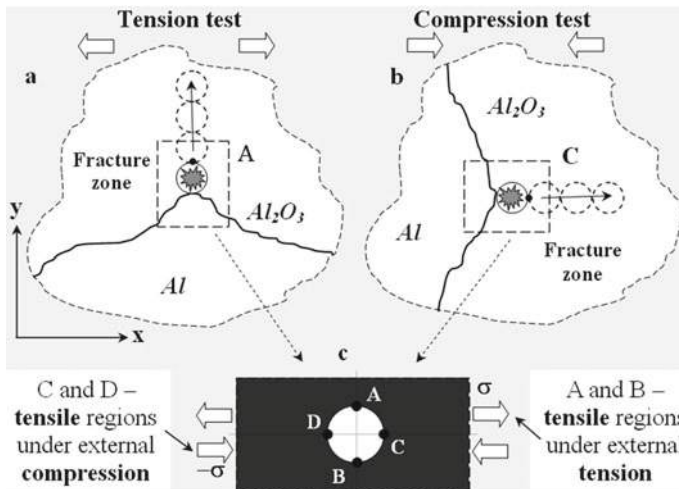


Fig. 35 Schematics of crack propagation in aluminum-alumina composite in different loading conditions

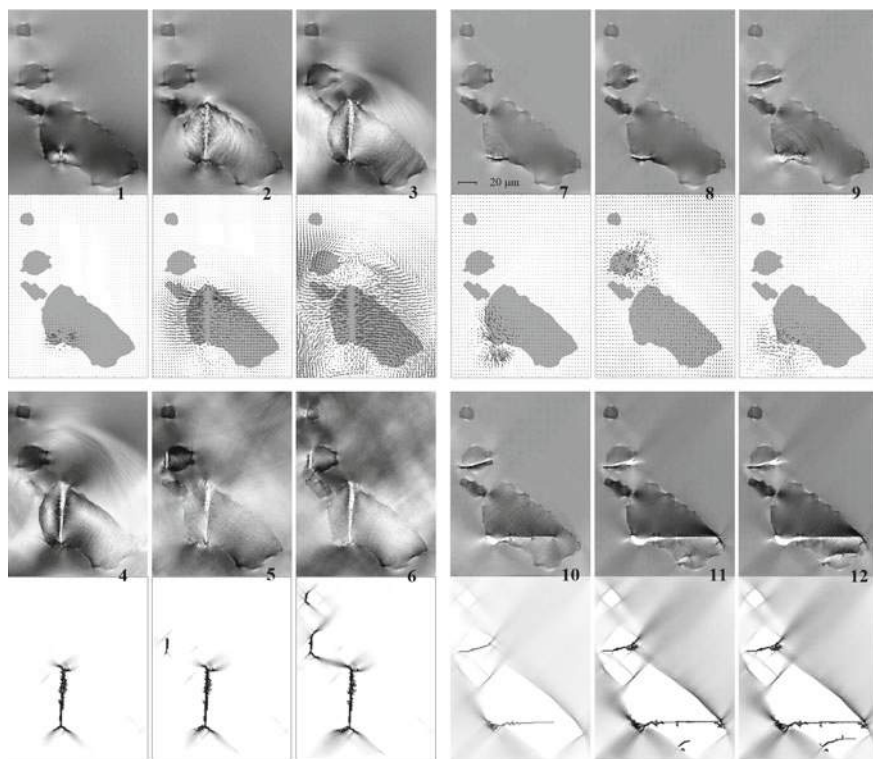


Fig. 36 Equivalent stress and plastic strain distributions and velocity fields in tension (1–6) and in compression (7–12) for the stress-strain curves in Fig. 34b

microstructure map are shown. The calculated results are presented for compression (1–6) and tension (7–12). The corresponding states (7–12) in the stress-strain curves are shown in Fig. 34b on an enlarged scale.

It is obvious from Fig. 36 (1) that in tension, the primary crack is initiated in the largest particle in the neighborhood of point A (Fig. 34a), where the stress concentration σ_{eq} is at its maximum. The crack propagates in the direction normal to that of tensile load and reaches the opposite side of the interface (Fig. 36 (1–4)). As this takes place, the adjacent regions are unloaded due to release waves propagating from newly formed free surfaces. This is supported by the descending portion of the stress-strain curve (Fig. 34(1–4)). The velocity fields in Fig. 36 (1–3) illustrate the crack opening process. Interactions of elastic waves with interfaces give rise to formation of a complex stressed state and may be responsible for generation of vortex structures. Rotations of local regions contribute to further increase in the stress concentration (Fig. 36 (3)). On further loading the average stress level in the region of interest is increased. Another stress concentrator is formed in a particle of smaller size, and new cracks are initiated (Fig. 36 (4–6)) with the resulting unloading of the material (Fig. 34 (2, 5, 6)). Figure 36 (4–6) shows the equivalent plastic strain distributions.

The fracture regions corresponding to a maximum value of plastic strain are shown in black. The tension crack nucleation and growth in the mesovolume are seen to occur essentially in the elastic deformation stage. This is attributed to the fact that the average stress level in the matrix is below the yield point $\sigma_0 = 62$ MPa (see Table 1), whereas the stress concentration in the local regions of the particles (points A in Fig. 34a) may be above a critical point $C_{ten} = 260$ MPa (see Table 2). Moderate plastic strain is seen in the neighborhood of points where local fracture zones are generated localized (Fig. 36 (1–4)).

In compression (Fig. 36 (7–12)), cracks are initiated at points of the maximum tensile stress concentration (points C in Fig. 34a). Notably, the equivalent stress at points of local compression (points A) is much higher than at points C. Thus, the compression fracture of the particle occurs at a much higher total stress level than that involved in tension (Fig. 34b) and is accompanied by high-intensity plastic deformation in the matrix (Fig. 36 (10–12)).

The compression cracks propagate in the loading direction. However, unlike the case of tensile load the crack propagation exhibits an oscillatory pattern. Severe plastic deformation of the matrix hinders fast increasing of the stress concentration. That is why the average velocity of compression-induced crack propagation through particles is much lower than in the case of tension-induced cracks. In consequence, cracking of the particles may occur in a switching mode. The regime in question can easily be traced by examining field velocities (Fig. 36 (7–9)). The first crack is initiated to produce an unloading effect of the mesovolume (Fig. 36 (7)). Then, the crack ceases to propagate, and the average stress level is seen to rise thereafter for a fairly long period of time due to strain hardening of the matrix (Fig. 34 (2, 7)). A new local fracture zone is formed in another particle. When a second crack ceases to grow the primary crack resumes its growth, propagating slowly (Fig. 34 (9–11)) to the opposite side of the interface (Fig. 36 (9–11)). Next, a third crack is formed (Fig. 36 (12)) and the process recurs.

In the calculations presented in Fig. 36, both for tension and for compression, the largest, medium-size, and the smallest particles are involved successively in the fracture process. Similar situations where the largest grains undergo the largest deformations and, in consequence, suffer the greatest damage have been observed experimentally. However, for an arbitrary microstructure of the type studied here (Fig. 34a), the sequence of events may be due to other reasons in addition to the size factor: different shape of the interface segments at points of crack nucleation and different types of the stressed-strained state of particles by virtue of their particular relative positions. We looked for ways to avoid the effects of the geometry and the loading conditions. To this end, a series of calculations were performed for composite microstructures subjected to tension. In the calculations, particles of the same shape and varying size (centers of masses of the particles) were aligned at the center of the computational domain. There are six independent combinations of relative positioning of particles according to this principle (Fig. 37). As in the former calculation presented in Fig. 36, the order of failure of the particles (the largest—medium-size—the smallest particles) is the same in all cases at hand. Such a fracture pattern is accounted for on the basis of the foregoing analysis of the calculated results

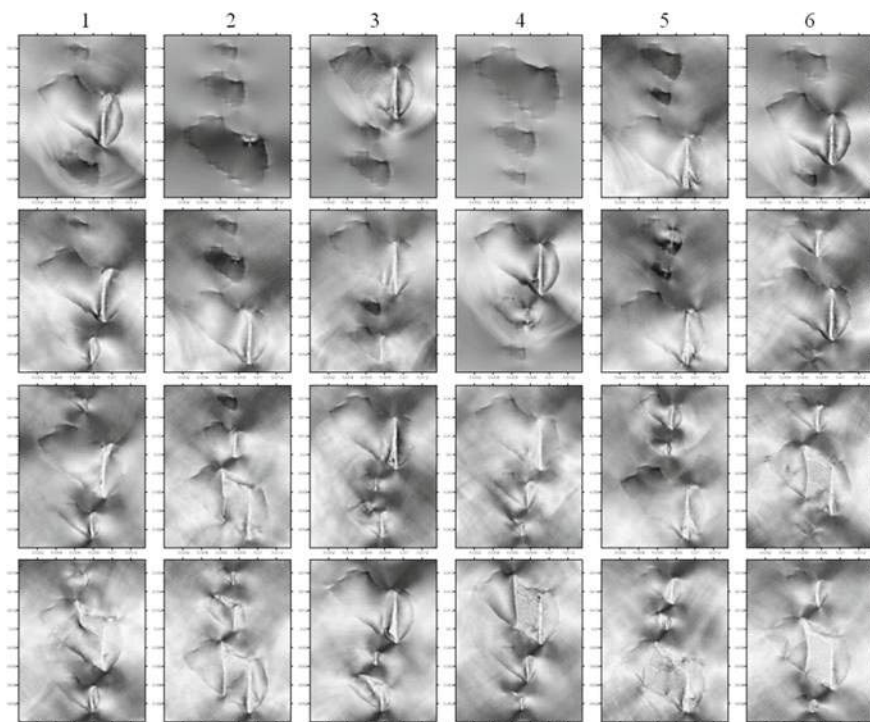


Fig. 37 Fracture patterns in composite mesovolumes with particles of the same shape and different size for 6 independent combinations of relative positions of the particles

and analytical estimation of the stress concentration: the larger the particle, the closer the stress concentration near the type II inhomogeneity approaches a maximum value obtained from an analytical solution for an infinite domain, and hence the sooner a local fracture zone is formed.

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Formation of a Nanostructured Hardened Surface Layer on the TiC-(Ni-Cr) Metal-Ceramic Alloy by Pulsed Electron-Beam Irradiation



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Abstract The efficiency and service life of products made from metal-ceramic tool alloys and used as cutting tools and friction units are determined by a combination of physical and strength properties of their surface layers with a thickness of up to 200 μm . Therefore, much attention is paid to their improvement at the present time. An effective way to increase the operational properties of the metal-ceramic alloy products is to modify the structure and the phase composition of the surface layers by forming multi-scale internal structures with a high proportion of low-dimensional (submicro and nano) components. For this purpose, surfaces are treated with concentrated energy fluxes. Pulse electron-beam irradiation (PEBI) in an inert gas plasma is one of the most effective methods. This chapter presents results of theoretical and experimental studies of this process. An example is the nanostructured hardened surface layer on the TiC-(Ni-Cr) metal-ceramic alloy (ratio of components 50:50) formed by PEBI in the plasma of argon, krypton, and xenon. Its multi-level structure, phase composition, as well as tribological and strength properties are shown.

Keywords Metal-ceramic composite · Pulsed electron-ion beam irradiation · Inert gas plasma · Structure · Wear resistance

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1 Introduction

Processing materials and products with concentrated energy fluxes to modify the structure, physical and strength characteristics of their surface layers is one of the effective ways to improve operational properties of metals and alloys. High-frequency [1–7], plasma [8–15], laser [16–23], ion- and electron-beam treatment [24–29] are widely used. The technological advantages of these methods include locality and non-contact heating of the surface layers. Each of these methods has its own features determined by the depth and the maximum temperature of the heated surface layer, heating and cooling rates, as well as duration and frequency of pulses. The purpose of the above surface hardening methods is to form multi-scale structures and phase compositions in the surface layer that determine a higher level of physical and strength properties. These structures and the phase compositions enable to decrease local plastic deformations and contribute to a more uniform distribution of elastic stresses in the material volume [30]. As a result, an energy level required to form stress concentrators in the nanostructured surface layer increases, and the probability of defects decreases. Also, a transition zone (instead of an interface) appears between the surface layer and the base material (they smoothly pass into each other). This determines the damping properties of the surface layer with respect to the base material under external shock mechanical and thermal loads. The transition zone prevents premature nucleation and propagation of microcracks from the surface into the material volume which cause its failure. The high-strength nanostructured surface layer, even having multiple micro-cracks, reduces the probability of the crack formation due to increased ductility.

Pulsed electron-beam irradiation (PEBI) has particular advantages for the modification of the structure and the phase composition of the surface layers of metals and alloys. It makes possible to treat a rather large surface area per one pulse with a power density of more than 10^6 W/cm^2 . Also, PEBI has some another benefits such as pulse frequency up to 10 s^{-1} , the low reflection coefficient from the irradiated metal surfaces (less than 10%) and high efficiency (more than 90%), as well as possibility to modify the surface layers up to several tens of micrometers in depth.

One of the first comprehensive studies of the metals treated by PEBI has been carried out by scientists of the Tomsk Scientific Center of the Siberian Branch of the USSR Academy of Sciences in the mid-1970s [31]. A little bit later, in the 1980s and subsequent years, some investigations have been done related to the impact of the high-energy electron beams on inorganic materials (Kurchatov Institute of Atomic Energy, Russia; D.V. Efremov Research Institute of Electrophysical Equipment, Russia; The Chernyshev Moscow Machine-Building Enterprise, Russia [30–34], and Sandia National Laboratory, USA [35, 36] have been devoted to amorphization and surface hardening of the materials. However, many important relationships and mechanisms of the formation of the phase composition and the substructure in the surface layer of the metals and the alloys after PEBI are still unclear. In particular, the modes that make possible to initiate the dynamic recrystallization and the

subsequent formation of nano- and submicroscale multiphase structures have not been established.

The mentioned possibilities to modify the surface layer of the metal and the alloys by PEBI are in many ways superior to other hardening technologies, such as deposition of wear-resistant coatings, ion irradiation and implantation, pulsed plasma and laser surface treatment [37–39]. The results of preliminary experimental studies have enabled to make conclusions that the development of an industrial technology for the modification of the surface layers is especially relevant for the manufacturing of metal-ceramic alloy tools. These alloys, based on refractory and high-hardness chemical compounds (carbides, nitrides, carbonitrides, and oxides) with a metal binder, are up-to-date dispersion-strengthened composite materials. They have high mechanical and tribological properties, such as strength, hardness, fracture toughness, wear resistance, and thermal stability. In this regard, the metal-ceramic alloys are widely used in various industries as functional materials for heavily loaded tribological units, machine parts, cutting elements operating under conditions of abrasive action, high temperatures and aggressive environments.

World-wide market analysis results show that about 67% of the total number of the metal-ceramic alloys is used for metal cutting tools, 13% is for drilling and rock cutting, 11% is for woodworking, and 9% is for chipless material-removal. At the same time, global consumption of the metal-ceramic alloy tools has doubled over the past 5 years.

Nowadays, the main direction of improving the operational properties of the metal-ceramic alloys is the development of technologies for sintering particles of highly hard components with a size of less than 1 μm (down to 100 nm). This increases the dispersion of the structure, hardness and strength of the alloys [40, 41]. Despite these advantages of the nanostructured hard alloys, their widespread implementation in the industries is hampered by the lack of effective technologies for mass production by the powder metallurgy method. Therefore, the widely used materials for the critical heavily loaded mechanisms are conventional metal-ceramic materials with particle sizes ranging from units to several tens of micrometers.

The operational efficiency of the metal-ceramic alloys in the mentioned conditions is determined by a set of mechanical properties of the surface layer with a thickness of 100–200 μm (the maximum permissible wear of a tool in the industries). Therefore, recent approaches to increase their lifecycle are based on the improvement of the surface layer characteristics. The main challenge is to eliminate powder sintering defects, including thermally activated micro- and macro-discontinuities of the solid structure. This is achieved by the interfacial interaction of the components under non-equilibrium thermodynamic conditions when heating to critical temperatures and then cooling at ultrahigh rates. The interaction of the metal and ceramic components of the alloys occurs at the peak of heating in the pulsed mode and, as a result, non-equilibrium structures are formed having new physical and strength properties. They remain in the surface layer after cooling at ultrahigh rate. A facility, designed and implemented in the Institute of High Current Electronics (Tomsk, Russia), makes it possible to modify the surface by PEBI [42]. The material is heated at a rate of up to 10^9 K/s and, then, cooled at a rate of $10^4 - 10^9$ K/s forming temperature gradients

Table 1 Atomic mass and ionization energy of the plasma-forming gases

Gas	Ionization energy (kJ/mol)	Atomic mass (g/mol)
Ar	1520.6	39.948
Kr	1350.0	83.798
Xe	1170.0	131.290

in the surface layer up to $10^7 - 10^8$ K/m [43]. To date, in collaboration with the Institute of Strength Physics and Materials Science SB RAS (Tomsk, Russia), a new approach has been developed to increase the depth of the surface layer on the metal-ceramic alloys which has a high content of nanoscale components. This is achieved by simultaneous pulsed irradiation with electron and ion beams in a single cycle. Inert gases (krypton and xenon), having higher atomic masses and lower ionization energy levels than argon (Table 1), are used as plasma-forming gas. Heavy ions interact with the surface of the irradiated material, increase the effective energy density in the electron beam, form a high level of radiation defects in the surface layer, and accelerate the processes of mass transfer and atom redistribution [43, 44].

This chapter presents the results of the studies of the nanostructured hardened surface layer on the TiC-(Ni-Cr) metal-ceramic alloy (ratio of components 50:50) formed by PEBI in the plasma of argon, krypton, and xenon. Its multi-level structure, phase composition, as well as tribological and strength properties are shown.

2 Temperature Fields in the Surface Layer under Pulsed Electron-Beam Irradiation

Temperature fields in the surface layer of the TiC-(Ni-Cr) metal-ceramic alloy heated by irradiation with high-energy beams of a rather large diameter can be quantified with a high degree of reliability using a one-dimensional model [45]. The model is based on PEBI of a cylindrical sample of radius r and length X from the end side. Denote electron-beam energy density E_s , pulse duration t_i , and pause between pulses t_0 . An approximation has been made in the model that the energy density distribution over the cross section of the electron beam is uniform. Therefore, the sample heating by the electron beam is determined by the equation [46]:

$$c\rho \frac{\partial T}{\partial t} = \lambda \frac{\partial^2 T}{\partial x^2} - \frac{\chi}{x_0} (T - T_0) - \frac{\delta}{x_0} (T^4 - T_0^4), \quad 0 \leq x \leq X, \quad (1)$$

where T is temperature; t is time; c , λ are heat capacity and thermal conductivity, respectively, in the solid material as a function of temperature; ρ is density; χ is the convective heat transfer coefficient; δ is the radiation heat transfer coefficient; T_0 is ambient temperature; x is longitudinal coordinate.

According to the additivity law, the heat capacity, thermal conductivity and density values of composite materials are calculated as:

$$\begin{aligned}
 c &= v_{Ni-Cr}c_{Ni-Cr} + v_{TiC}c_{TiC}, \lambda = v_{Ni-Cr}\lambda_{Ni-Cr} + v_{TiC}\lambda_{TiC}, \\
 \rho &= v_{Ni-Cr}\rho_{Ni-Cr} + v_{TiC}\rho_{TiC}, c_{Ni-Cr} = (c_{Ni}a_{Ni} + c_{Cr}a_{Cr})/(a_{Ni} + a_{Cr}), \\
 c_{TiC} &= (c_{Ti}a_{Ti} + c_Ca_C)/(a_{Ti} + a_C), \\
 \lambda_{Ni-Cr} &= (\lambda_{Ni}a_{Ni} + \lambda_{Cr}a_{Cr})/(a_{Ni} + a_{Cr}); \lambda_{TiC} = (\lambda_{Ti}a_{Ti} + \lambda_Ca_C)/(a_{Ti} + a_C), \\
 \rho_{Ni-Cr} &= (\rho_{Ni}a_{Ni} + \rho_{Cr}a_{Cr})/(a_{Ni} + a_{Cr}); \rho_{TiC} = (\rho_{Ti}a_{Ti} + \rho_Ca_C)/(a_{Ti} + a_C),
 \end{aligned} \tag{2}$$

where c_j , λ_j , ρ_j , v_j are the heat capacity, thermal conductivity, density and relative mass fraction of the j -th component of the composite (the TiC-(Ni-Cr) metal-ceramic alloy in the investigated case).

For a single electron pulse, boundary conditions on the irradiated surface ($x = 0$) are represented as:

$$-\lambda \frac{\partial T}{\partial x} = \begin{cases} E_s/t_i > 0, t \in [(n-1)(t_i + t_0), (n-1)t_0 + nt_i] \\ 0, t \notin [(n-1)(t_i + t_0), (n-1)t_0 + nt_i] \end{cases}, x = X : \frac{\partial T}{\partial x} = 0 \tag{3}$$

where n is the number of pulses.

The initial data for the TiC-(Ni-Cr) metal-ceramic alloy having the ratio of the components 50:50 are [47–49]: $c_{NiCr} = 452 \text{ J/kgK}$, $c_{TiC} = 408 \text{ J/kgK}$, $\lambda_{NiCr} = 88.5 \text{ J/sKm}$, $\lambda_{TiC} = 42 \text{ J/sKm}$, $\rho_{NiCr} = 8800 \text{ kg/m}^3$, $v_{NiCr} = v_{TiC} = 0.5$, $\chi = 10 \text{ J/sKm}^2$, $\delta = 3 \times 10^{-7} \text{ J/sK}^4\text{m}^2$, $T_0 = 300 \text{ K}$, $r = 0.01 \text{ m}$, $x = 0.001 \text{ m}$.

An increase in pulse duration causes a significant decrease in temperature of the surface layer and an increase in its depth. When pulse duration is $200 \mu\text{s}$, the surface layer temperature achieves 550 K and its depth slightly increases. The electron-beam energy density values have a strong effect on the temperature profile in the surface layer. Both temperature and depth of the surface layer rise simultaneously with an increase in electron-beam energy density. For example, the surface layer temperature rises from 760 to 1250 K when electron-beam energy density enhances from 5 to 10 J/cm^2 . Its further increase up to 40 J/cm^2 raises the surface layer temperature up to 4250 K . This effect becomes stronger with increasing pulse duration.

At a constant energy density value, an increase in the number of pulses up to 300 enhances the heated surface layer depth by more than two times. Then, upon reaching a certain number of pulses, the surface layer depth remains constant rising only with an increase in electron-beam energy density. At a frequency of 1 s^{-1} , an increase in the number of pulses up to 50 has almost no effect on the surface layer temperature which increases only after 100 pulses. The temperature profile in the surface layer changes little with increasing pulse frequency from 1 to 20 s^{-1} .

The temperature gradient under PEBI significantly affects structural and phase transformations in the surface layer. The calculated dependences of the temperature

gradient in the surface layer of the TiC-(Ni-Cr) metal-ceramic alloy from pulse frequency and duration, the number of pulses, as well as electron-beam energy density show that an increase in pulse duration sharply decreases the temperature gradient while, on the contrary, an increase in electron-beam energy density enhances it. Rising pulse frequency and the number of pulses do not affect the temperature gradient in the surface layer. Thus, the main patterns of the structural and phase transformations in the surface layer of the TiC-(Ni-Cr) metal-ceramic alloy under PEBI are determined primarily by electron-beam energy density, as well as the number of pulses and pulse duration. A weak correlation of the temperature distribution in the surface layer with the number of pulses and pulse frequency means that the temperature profiles are formed during the first pulse and remains almost unchanged with an increase in these parameters. Raising the number of pulses enhances duration of the surface layer existence in non-equilibrium temperature–time conditions. It also determines the interfacial interaction of the surface layer components and, as a result, the formed structure.

An analysis of the temperature profiles and the surface layer depth enables to estimate the ranges of the experimental parameters (electron-beam energy density, the number of pulses and pulse duration) for the surface layer structure modification by PEBI. The main criteria are the modified surface layer depth (from 100 to 200 μm), an increase in its temperature during the first pulse (up to 3000 K), and a minimum temperature gradient. Electron-beam energy density of 40 – 50 J/cm² and pulse duration of 100–200 μs correspond to these criteria. Taking into account that the temperature profiles in the surface layer are formed precisely during the first pulse and remain almost unchanged then, an increase in the number of pulses changes only the duration of the interfacial interaction of the components in the TiC-(Ni-Cr) metal-ceramic alloy under non-equilibrium temperature–time conditions for the specified electron-beam energy density and pulse duration values. Thus, the results of the surface layer temperature profile calculations enable to verify their compliance with the general patterns. However, an increase in the content of the ceramic component in the TiC-(Ni-Cr) metal-ceramic alloy enhances the maximum heating temperature in the surface layer, as well as the temperature gradient in the surface layer.

3 The Effect of Pulsed Electron-Beam Irradiation in Different Plasma-Forming Gases on the Surface Layer Structure and Properties

3.1 Material and Experimental Methods

Samples (10 × 10 × 4 mm plates) were made from the TiC-(Ni-Cr) metal-ceramic alloy. The flat sample surfaces were polished to a mirror shine before irradiation. PEBI was performed using a plasma cathode setup [50, 51]. Argon, krypton, and xenon were used as a plasma-forming gas; their atomic mass and ionization energy

Table 2 PEBI parameters

No.	Plasma-forming gas	Pulse duration (μs)	Electron-beam energy density (J/cm^2)	Number of pulses
1	Ar	200	40	15
2	Kr	100	40	15
3	Xe	100	40	15
4	Ar	150	60	15
5	Kr	200	60	15
6	Xe	150	60	15

are presented in Table 1. The values of electron-beam energy density, pulse duration, and frequency are presented in Table 2. The wide-aperture electron beam covered the entire sample surfaces.

Specimens for the microstructure analysis were prepared by the focused ion beam method using an ‘Jeol EM-09100IS’ setup (accelerating voltage 8 kV, ion beam incidence angle 4°).

Three-point bending tests were carried out using an ‘Instron-3369’ machine under continuous loading at a speed of 0.2 mm/min in accordance with ISO 3327-82. Transverse bending strength was calculated without taking into account any possible plastic deformations by the formula $\sigma = 3Pl/2h^2b$, where P was force corresponding to the sample failure; l was the distance between the pillars; h was the sample height (the size coincided with the force direction); b was the sample width (perpendicular to height).

X-ray phase analysis of the surface layers of the samples having the minimum and maximum values of transverse bending strength after PEBI in different plasma-forming gases was carried out using a ‘Shimadzu XRD 6000’ X-ray diffractometer in Cu radiation (accelerating voltage 40 kV, current 30 mA).

Microhardness values on the surface layers were measured using a ‘DM 8B’ microhardness tester in accordance with ISO 14577-1:2015. Nanohardness and Young’s modulus of the surface layers were investigated using a ‘CSEM Nano Hardness Tester’ facility in accordance with ISO 14577-1:2015.

Wear resistance of the surface layers and the friction coefficients on the sample surfaces were studied using a ‘CSEM Tribometer High Temperature S/N 07-142’ tribometer (CSEM Instruments, Switzerland). The samples were rotated on the stationary counterpart (a diamond cone) during the tests. Load on the diamond cone was 5 N, the number of the sample revolutions was 2500. Friction force was continuously measured using a ‘Micromesure System STIL’ micrometer system (Science et Techniques Industrielles de la Lumiere. STILS. A, France). Then, absolute friction coefficient values were calculated. The surface profiles (the depth and the cross-section area of grooves made with the diamond cone) were identified using a ‘MICRO MEASURE 3D station (STIL)’ profiler after the end of the tests.

Also, abrasive wear resistance of the sample surfaces was investigated in accordance with ASTM G65. The flat samples were pushed down to a rotated rubber

lined wheel with a diameter of 218 mm. Load was 36 N, rotating speed was 200 revolutions per minute. An abrasive powder (13A electrocorundum with a particle size of $200 - 250 \mu\text{m}$) was supplied to the friction surface. The abrasive flow rate was $270 - 280 \text{ g/min}$.

3.2 *Changes in the Structure and the Properties of the Surface Layer after Pulsed Electron-Beam Irradiation*

The study of the TiC-(Ni-Cr) metal-ceramic alloy microstructure by scanning electron microscopy (SEM) showed that titanium carbide particles in the initial state (after sintering) had pronounced unequal shapes (Fig. 1a). The average particle size was $2.7 \pm 1.2 \mu\text{m}$, the maximum size was $7.6 \mu\text{m}$ (Fig. 1b). The alloy in the initial state had three main structural components:

- (1) high-strength titanium carbide particles with a size of 1–10 microns;
- (2) interparticle layers of the metal binder;
- (3) “particles–binder” transition zones with a width of about four microns.

Fracture surfaces of the samples had a brittle intergranular appearance (Fig. 2). Figure 3 shows SEM images of the alloy structure in the initial state, as well as titanium carbide particles and the nickel-chrome binder.

After PEBI, the shapes of titanium carbide particles smoothed and became more rounded (especially of small ones) in the surface layer (Fig. 4a). In addition, their average sizes changed (Fig. 4b). As electron-beam power density (W_S) increases, their average and maximum sizes changed along curves with a maximum at $3.2 \times 10^5 \text{ W/cm}^2$ (Fig. 5, curve 1 and 2, respectively).

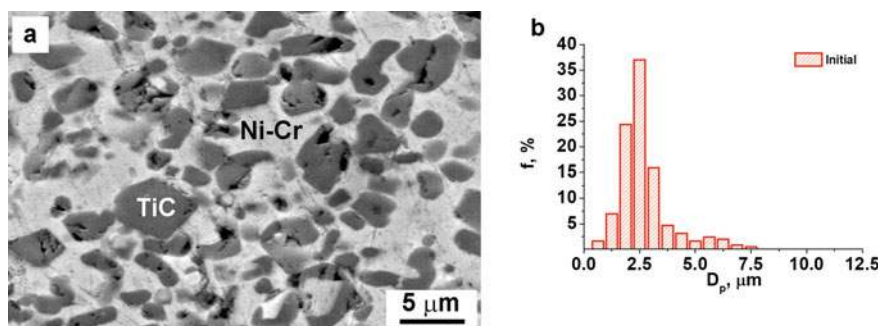


Fig. 1 SEM images of the structure of the metal-ceramic alloy polished surface in the initial state (a) and the element distribution in the TiC-(Ni-Cr) metal-ceramic alloy at the interface between titanium carbide particles and the nickel-chrome binder (b)

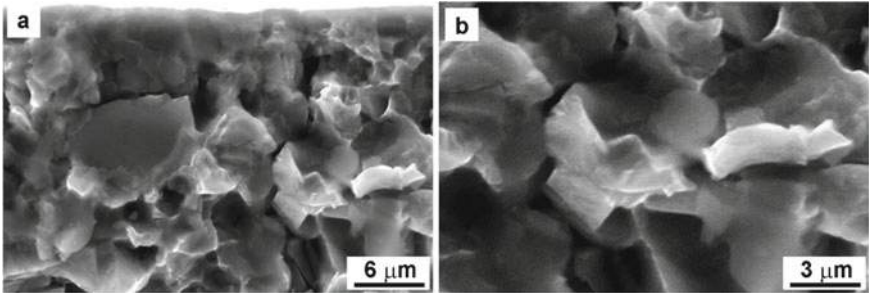


Fig. 2 The fracture surface of the TiC-(Ni-Cr) metal-ceramic alloy in the initial state

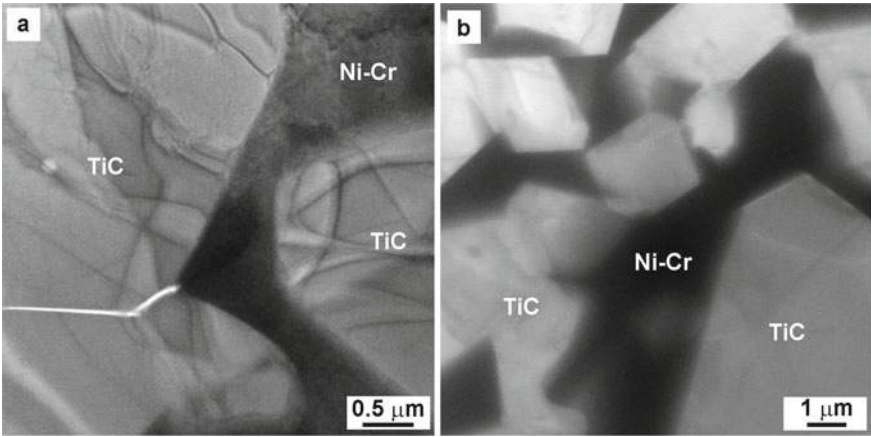


Fig. 3 TEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy in the initial state

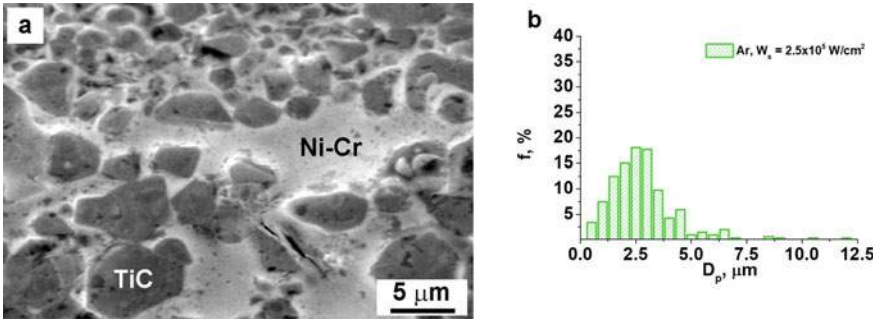


Fig. 4 The structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI ($W_s = 2.5 \times 10^5 W/cm^2$) in the argon plasma (a) and the size distribution of titanium carbide particles (b, $D = 2.8 \pm 1.7 \mu m$)

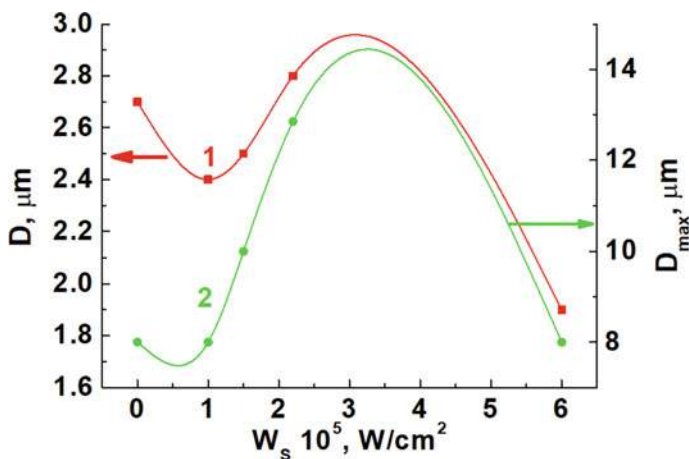


Fig. 5 The average (curve 1) and maximum (curve 2) titanium carbide particle sizes on the irradiated surface versus electron-beam power density

PEBI caused high-rate melting and subsequent solidification of the nickel-chrome binder in the surface layer. As a result, titanium carbide particles were refined by partially dissolving in the molten nickel-chrome binder and simultaneous large particle cracking due to high thermal stresses (Figs. 6 and 7).

In addition to microcracking, PEBI changed the structure and the phase composition of the surface layer. The next process options occurred depending on the PEBI parameters: the nickel-chrome binder melted and solidified, or the primary titanium carbide particles partially dissolved and the secondary titanium carbide precipitated, or large carbide particles cracked and these microcracks were filled with the molten nickel-chrome binder. Figures 8 and 9 show microphotographs of the surface layer after PEBI that demonstrate the results of the mentioned processes.

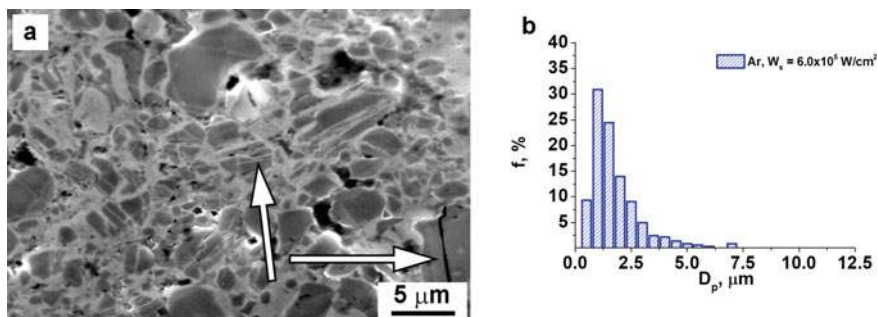


Fig. 6 The structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI ($W_s = 6.0 \times 10^5 \text{ W/cm}^2$) in the argon plasma (a), and the distribution of the titanium carbide particle sizes (b, $D = 1.9 \pm 1.2 \mu\text{m}$)

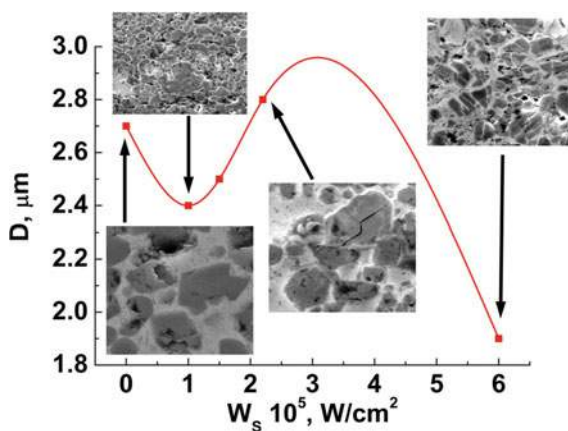


Fig. 7 Failure types of titanium carbide particles under thermo-elastic stresses versus electron-beam power density after irradiation in the argon plasma

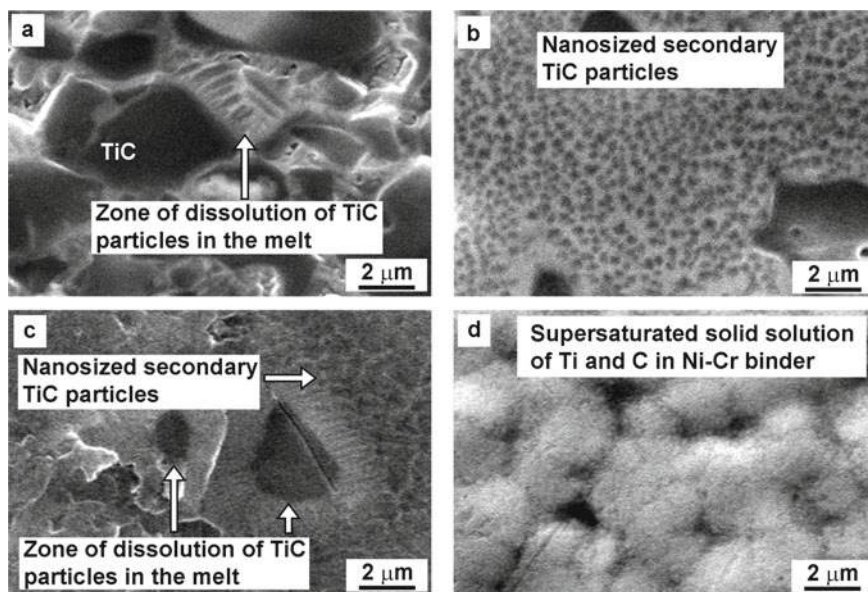


Fig. 8 The surface structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI in the argon plasma at an electron-beam power density from 2×10^5 up to 10×10^5 W/cm² : *a* – $E_S = 40.0$ J/m², $t_i = 200.0$ μs, $W_S = 2 \times 10^5$ W/cm²; *b* – $E_S = 40.0$ J/cm², $t_i = 80.0$ μs, $W_S = 4 \times 10^5$ W/cm², *c* – $E_S = 40.0$ J/cm², $t_i = 50.0$ μs, $W_S = 8 \times 10^5$ W/cm²; *d* – $E_S = 2.5$ J/cm², $t_i = 2.5$ μs, $W_S = 10 \times 10^5$ W/cm²

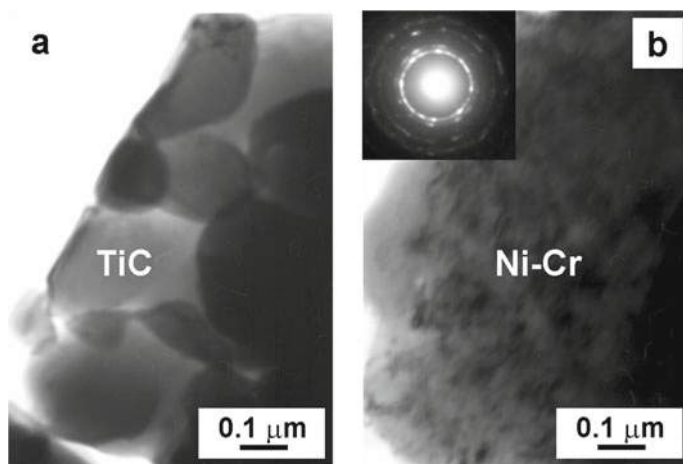


Fig. 9 TEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI ($E_S = 40 \text{ J/cm}^2$, $t_i = 50 \mu\text{s}$, $V = 1 \text{ s}^{-1}$, $n = 15$ pulses): **a**—titanium carbide particles; **b**—the nickel-chrome binder

The various stages of the interaction of titanium carbide particles with the molten nickel-chrome binder are illustrated by the micrographs in Fig. 8. The initial stage of the titanium carbide particle melting was at $2 \times 10^5 \text{ W/cm}^2$ (Fig. 8a). Then, the sizes of the primary titanium carbide particles and their volume fraction decreased with increasing electron-beam power density (Fig. 8b, c). Finally, the process completed with the partial dissolution of titanium carbide particles in the molten nickel-chrome binder and the formation of a nanostructure in the surface layer (Fig. 8d).

The partial dissolution of the primary titanium carbide particles and the formation of titanium carbide dissolution zones in the molten nickel-chrome binder occurred in the surface layer at an electron-beam power density of $2 \times 10^5 \text{ W/cm}^2$ (Fig. 8a). Rising electron-beam power density up to $4 \times 10^5 \text{ W/cm}^2$ caused an increase in dissolution rate of the primary titanium carbide particles in the molten nickel-chrome binder, the formation of the extensive titanium carbide dissolution zones, and the release of the secondary titanium carbide nanoparticles inside them (Fig. 8b). A subsequent increase in electron-beam power density up to $8 \times 10^5 \text{ W/cm}^2$ resulted in the almost complete dissolution of the primary titanium carbide particles in the surface layer, and the subsequent formation of a supersaturated titanium-carbon solid solution in the nickel-chromium binder at $10 \times 10^5 \text{ W/cm}^2$ (Fig. 8c, d).

A glass-like structure formed in the surface layer. In this structure, titanium carbide particles were almost completely dissolved in the molten nickel-chrome binder (Fig. 9). Microdiffraction studies showed that the glass-like structure consisted of the submicron (200–250 nm) titanium carbide particles and the nanostructured nickel-chrome binder (crystallite size was in the range of 10–30 nm).

Figure 10 shows typical surface layer structures after PEBI using various pulse durations with constant other energy parameters (electron-beam energy density was

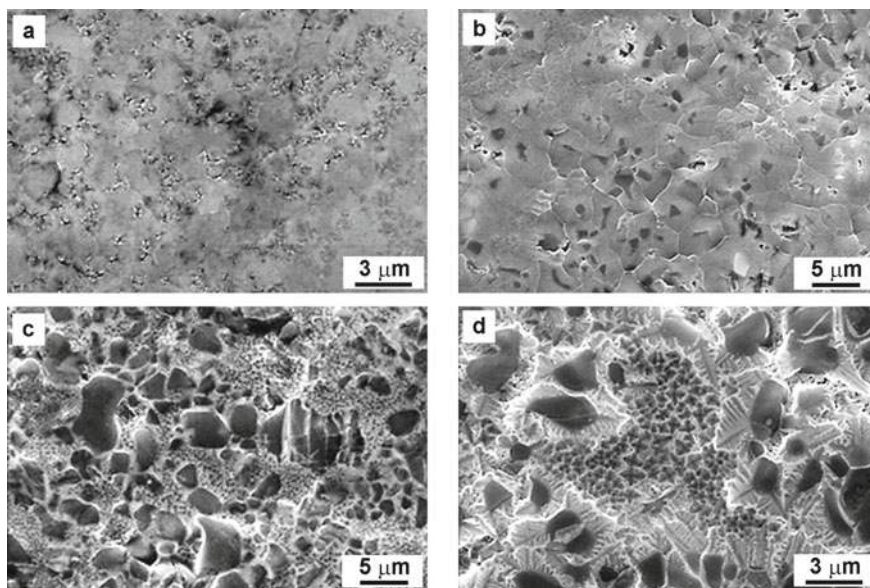


Fig. 10 SEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI ($E_s = 40 \text{ J/cm}^2$, $v = 1 \text{ s}^{-1}$, $n = 15$ pulses): **a** $t_i = 50 \mu\text{s}$, **b** $t_i = 100 \mu\text{s}$, **c** $t_i = 150 \mu\text{s}$, **d** $t_i = 200 \mu\text{s}$

40 J/cm^2 , pulse frequency was 1 s^{-1} , the number of pulses was 15). Almost complete dissolution of the primary titanium carbide particles occurred in the molten nickel-chrome binder, and a titanium-carbon solid solution was formed in the surface layer at a pulse duration of $50 \mu\text{s}$. An increase in pulse duration up to $100 \mu\text{s}$ caused the formation of a fine-grain structure with residues of the primary titanium carbide particles in the central part of each grain. The uniform structure was formed at a pulse duration of $150 \mu\text{s}$. It included the primary titanium carbide microparticles and interlayers of the nickel-chrome binder saturated with the secondary titanium carbide nanoparticles.

The dependence of the dimension of the titanium carbide particles on electron-beam power density is shown in Fig. 11. At low electron-beam power density values, large titanium carbide particles were refined due to thermal shock and partially dissolved in the molten nickel-chrome binder. Then, this effect intensified to a large extent with an increase in electron-beam power density. As a result, a supersaturated titanium-carbon solid solution in the nickel-chrome binder was formed at electron-beam power density of $8 \times 10^5 \text{ W/cm}^2$ and higher.

Finally, the secondary titanium carbide nanoparticles precipitated in the nickel-chrome binder upon cooling (Fig. 12).

It was found by the local X-ray spectral analysis that the titanium concentration in the nickel-chrome binder interlayers varied within 19–29 wt% and increased

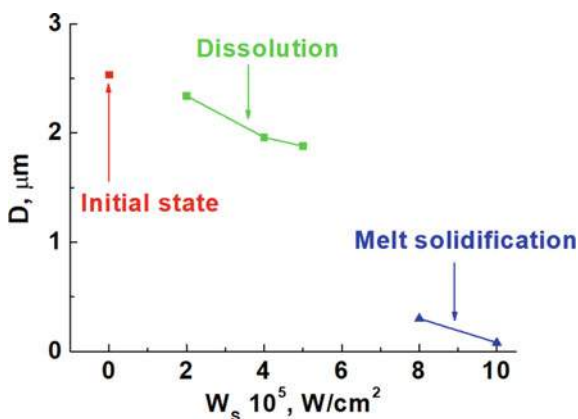


Fig. 11 Average titanium carbide particle sizes versus electron-beam power density

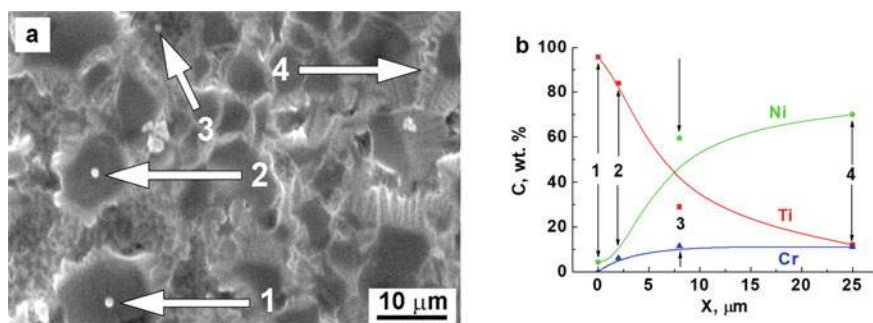


Fig. 12 The structure of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI (a, $E_S = 40 \text{ J/cm}^2$, $t_i = 200 \mu\text{s}$, $v = 1 \text{ s}^{-1}$, $n = 15$ pulses), and the effect of the distance from titanium carbide particles on the nickel, titanium, and chromium content (b, numbers indicate the measurement points and the results of local quantitative X-ray spectral analysis)

significantly near the interface between titanium carbide particles and the nickel-chrome binder. The values were maximum at the surfaces of titanium carbide particles (Fig. 13). These results indicated the intensive dissolution of titanium carbide particles in the molten nickel-chrome binder and its excessive saturation with titanium and carbon, as well as proved the inevitability of the formation of the multiphase composition in the surface layer during its high-rate cooling after electron-beam pulses. In this regard, it was suggested that one of the reasons for the nanostructure (Fig. 10a) formation in the surface layer after PEBI ($E_S = 40 \text{ J/cm}^2$, $t_i = 50 \mu\text{s}$ and 15 pulses) was the supersaturation of the nickel-chromium binder with carbon and titanium.

These features were revealed by the results of the SEM investigations (Fig. 13). Dendritic structures typical for high-rate solidification were formed at the mentioned PEBI parameters. However, there were some differences depended on the PEBI mode due to various heat input rates characterized by electron-beam power density.

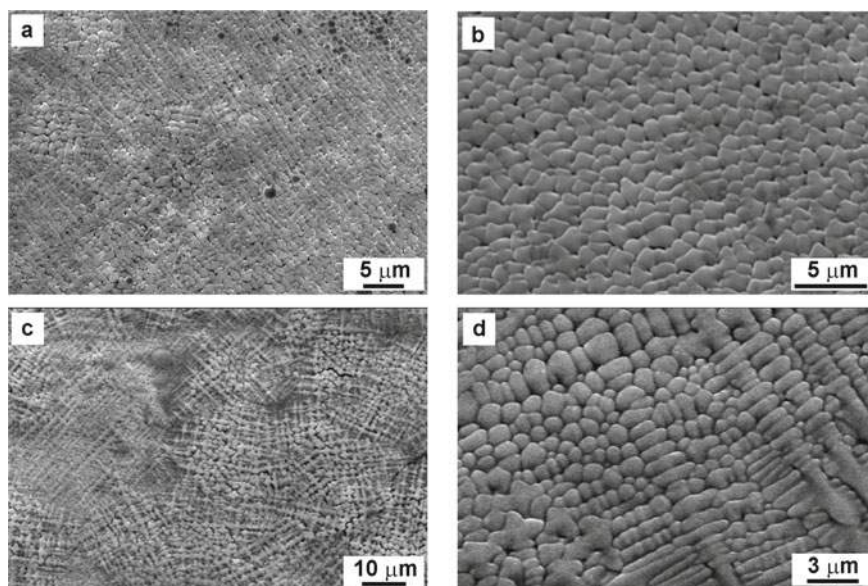


Fig. 13 SEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy surface after PEBI: **a** and **b** – $E_S = 40 \text{ J/cm}^2$, $t_i = 50 \mu\text{s}$, $n = 15$ pulses; **c** and **d** – $E_S = 40 \text{ J/cm}^2$, $t_i = 200 \mu\text{s}$, $n = 15$ pulses

In the formed structure, titanium carbide particles with a size of $0.5 - 1.5 \mu\text{m}$ had an almost equiaxial shape at a pulse duration of $50 \mu\text{s}$ (Fig. 13a, b). An increase in pulse duration up to $200 \mu\text{s}$ decreased cooling rate of the surface layer. Therefore, a dendritic structure with clearly defined first and second order axes was formed during solidification (Fig. 13c, d). Carbide particle shapes varied from an isotropic type with a size of $0.5 - 1.5 \mu\text{m}$ to a highly anisotropic one having a length of $1.5 - 6 \mu\text{m}$ and a width of $0.5 - 0.8 \mu\text{m}$.

Thus, the phase composition, the defective substructure state, and the solid solution in the gradient structure formed by PEBI depended on the location of the initial structure with respect to titanium carbide particles. The whole thickness of the modified surface layer increased from 13 to $40 \mu\text{m}$ with rising pulse duration from 50 to $200 \mu\text{s}$, but the thickness of its glass-like part continuously decreased down to zero at a pulse duration of $200 \mu\text{s}$.

Modification of the structure and the phase composition in the surface layer changed the mechanism of its failure. Fractography analysis of the surface layer after PEBI with a pulse duration of $200 \mu\text{s}$ and an electron-beam energy density of 40 J/cm^2 (Fig. 14) showed that failures had occurred by the ductile-brittle mechanism (numerous plastic deformation steps were on the facets) while the initial sample fracture type was brittle.

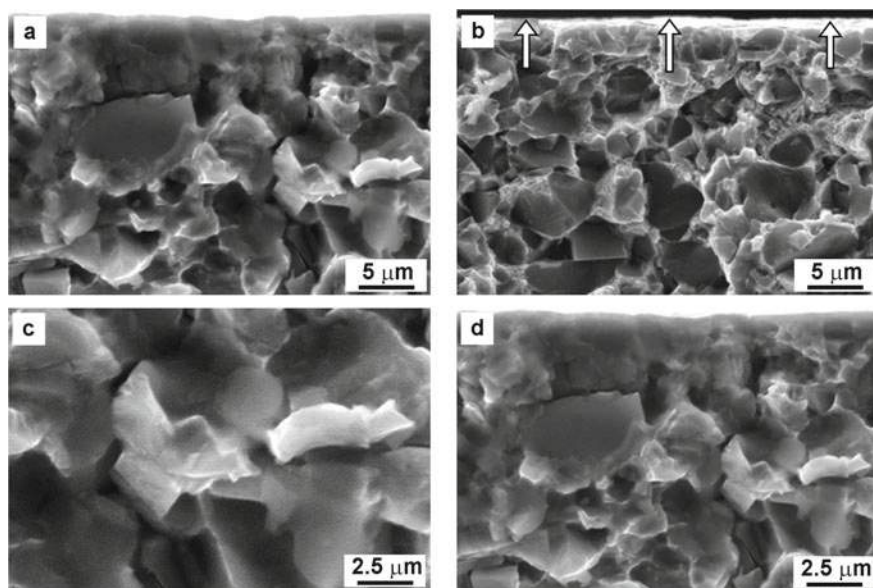


Fig. 14 The fracture surface of the TiC-(Ni-Cr) metal-ceramic alloy in the initial state (a and b, the arrows indicate the irradiation surface) and after PEBI (c and d, $E_S = 40 \text{ J/cm}^2$, $t_i = 200 \mu\text{s}$, $n = 15$ pulses)

The revealed diversity of the structures and the phase composition in the surface layer after PEBI was due to a high level of temperature gradients and the distribution of the alloying elements. The basic indication of the non-equilibrium state achievement in the surface layer was the formation of the modified structure included titanium carbide submicroparticles and the nanostructured nickel-chrome binder. The nickel-chrome binder nanostructuring changed the fracture mechanism from the brittle type to the ductile-brittle one. In this case, the brittle component was determined by the titanium carbide particle failures, and the ductile one was driven by the failure of the nanostructured nickel-chrome binder.

3.3 *Theoretical Assessment of the Effect of Plasma-forming Gases on the Pulsed Electron-Beam Irradiation Process*

An integral part of the PEBI process is the plasma formation by the inert gas ionization in a vacuum chamber. The inert gas plasma is also an electron source in the case of using a plasma-cathode setup. Additionally to electrons, inert gas ions interact with the irradiated material surface and enhance the surface layer modification. Argon is typically used as a plasma-forming gas for PEBI. However, krypton and xenon are

also of significant interest. They differ significantly from argon in terms of atomic mass [52] and ionization energy [53] (Table 1).

It can be assumed that atomic mass and ionization energy of a plasma-forming gas affects the result of the surface layer modification by PEBI at the same pressure in the electron-beam setup chamber. In order to verify or disprove this, PEBI of the end surface of a round plate TiC-(Ni-Cr) metal-ceramic alloy sample with a radius of r and a thickness of X is considered. When the energy density distribution in the electron-beam cross section is uniform, the thermal-conductivity equation is similar to the Eq. (1).

For a single electron pulse, boundary conditions on the irradiated surface ($X = 0$) are represented as:

$$-\lambda \frac{\partial T}{\partial x} = \begin{cases} E_S(g)/t_i > 0, & 0 < t \leq t_i \\ 0, & t > t_i \end{cases}, \quad (4)$$

and at the end ($x = X$)

$$-\lambda \frac{\partial T}{\partial x} = \chi(T - T_0) + \varepsilon(T^4 - T_0^4) \quad (5)$$

In Eqs. (4) and (5), the following notations are used: t_i is pulse duration; $E_S(g)$ is effective electron-beam energy density in an inert gas plasma. In order to determine $E_S(g)$, it was assumed that an additional ion flux, formed in the inert gas plasma due to electron-beam energy exposure, increases effective (total) radiation density. It is presented in the following approximated form:

$$E_S(g) \approx E_S(Ar) + \Delta E(g) \quad (6)$$

where $E_S(Ar)$ is electron-beam energy density in the argon plasma, $\Delta E(g)$ is electron-beam energy density increment due to additional ionization in the krypton or xenon plasma ($g = Kr$ or Xe). It is also suggested that a decrease in ionization energy of the inert gas increases the number of ions in the electron-plasma flow and, accordingly, the value of $\Delta E(g)$. Hence, effective electron-beam energy density increases with decreasing gas ionization energy and increasing $\Delta E(g)$. Since $E_{iAr} > E_{iKr} > E_{iXe}$ (according to Table 1), the next condition $0 < \Delta E(Kr) < \Delta E(Xe)$ is satisfied, which can be rewritten as

$$E_S(Ar) < E_S(Kr) < E_S(Xe) \quad (7)$$

Numerical calculations were carried out using the system of the Eqs. (2)–(5) and the relations (6) and (7). Heat capacity and thermal conductivity of the TiC-(Ni-Cr) metal-ceramic alloy were taken constant. The values of thermo-physical properties are represented in paragraph 2 and in [54–58].

In order to facilitate the analysis of the effect of the plasma-forming gas on the possibility and intensity of the structure and phase transformations in the surface

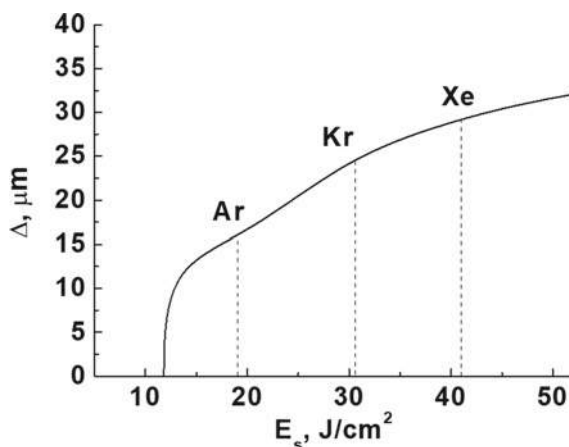


Fig. 15 Depth of the TiC-(Ni-Cr) metal-ceramic alloy surface layer heated up to a temperature of 1500 K by PEBI (argon, krypton, and xenon plasma, $t_i = 150 \mu\text{s}$, $n = 1$ pulse) versus energy density

layer under PEBI, the characteristic temperature index T^* has been used. Structural and phase transformations are negligible when $T < T^*$, and are quite intense when $T \geq T^*$. Figure 15 shows the dependence of the surface layer depth heated up to $T^* = 1500 \text{ K}$ under single electron-beam pulse with a duration of $150 \mu\text{s}$ on effective electron-beam energy density for argon, krypton and xenon used as a plasma-forming gas. It increases with the change of the plasma-forming gas from argon to krypton, and, then, to xenon at constant pulse duration and electron-beam energy density. Accordingly, there is also a high probability of an increase in the intensity of the interfacial interaction of the TiC-(Ni-Cr) metal-ceramic alloy components in the surface layer and its depth.

3.4 The Effect of the Plasma-Forming Gases on the Structure and the Properties of the Modified Surface Layer

The results of the SEM studies of the modified surface layer structure after PEBI in the different plasma-forming gases showed their patterns in terms of heterogeneity and diversity in the quantitative structure parameters. All modified layers consisted of a set of sublayers. The top sublayer, directly adjacent to the irradiated surface, had a nanosized metal-ceramic columnar structure oriented perpendicular to the outer sample surface (Figs. 16 and 17, sublayer 1). Below, a thicker sublayer was formed with a coarser columnar metal-ceramic structure (Figs. 16 and 17, sublayer 2). Columns were mostly oriented perpendicular to the outer sample surface. The next sublayer had a dendritic type structure. It was a transition layer to the initial metal-ceramic structure (Figs. 16 and 17, sublayer 3). The thicknesses of the mentioned

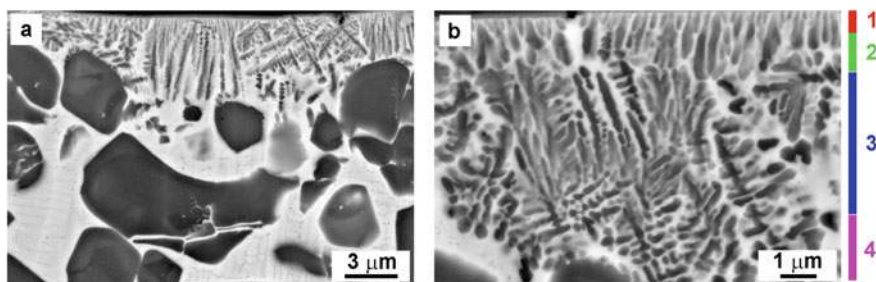


Fig. 16 SEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy surface layer after PEBI (argon plasma, $E_S = 60 \text{ J/cm}^2$)

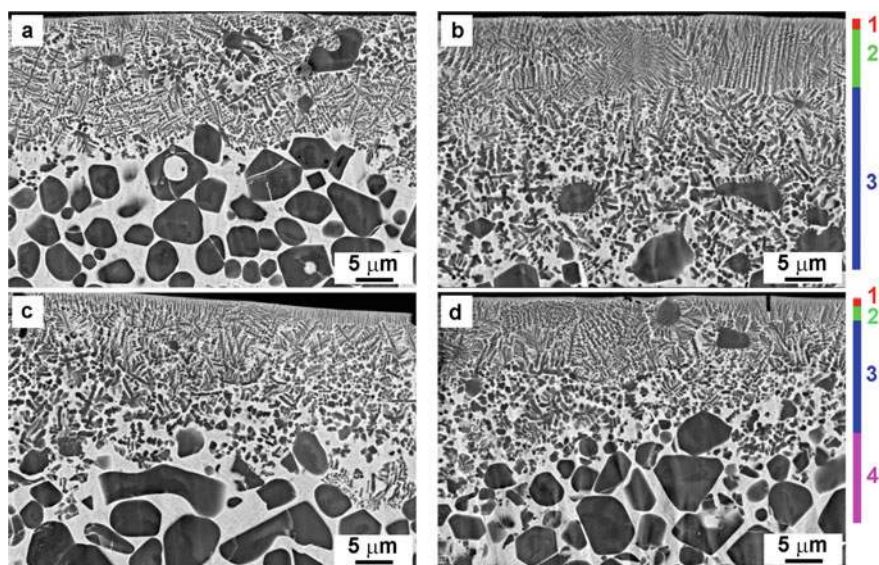


Fig. 17 SEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy surface layer after PEBI: **a** and **b**—krypton plasma, $E_S = 40 \text{ J/cm}^2$; **c** and **d**—xenon plasma, $E_S = 60 \text{ J/cm}^2$

sublayers depended on the PEBI parameters and the plasma-forming gas. Figure 18 shows the dependences of the thickness of the top sublayer and the whole modified surface layer on pulse duration at electron-beam energy densities of 40 and 60 J/cm² using argon, krypton, and xenon as a plasma-forming gas. The effect of the electron-beam energy density on the whole modified surface layer thickness was minimal. However, it significantly increased in the krypton plasma, and even more in the xenon one. The range of the maximum thicknesses of the modified surface layer varied from 35 μm in the argon plasma to 40 μm in the krypton one, and 47 μm in the xenon plasma at an electron-beam energy density of 60 J/cm² (Fig. 18). An increase in pulse duration caused an increase in the thickness of both the top sublayer and the whole

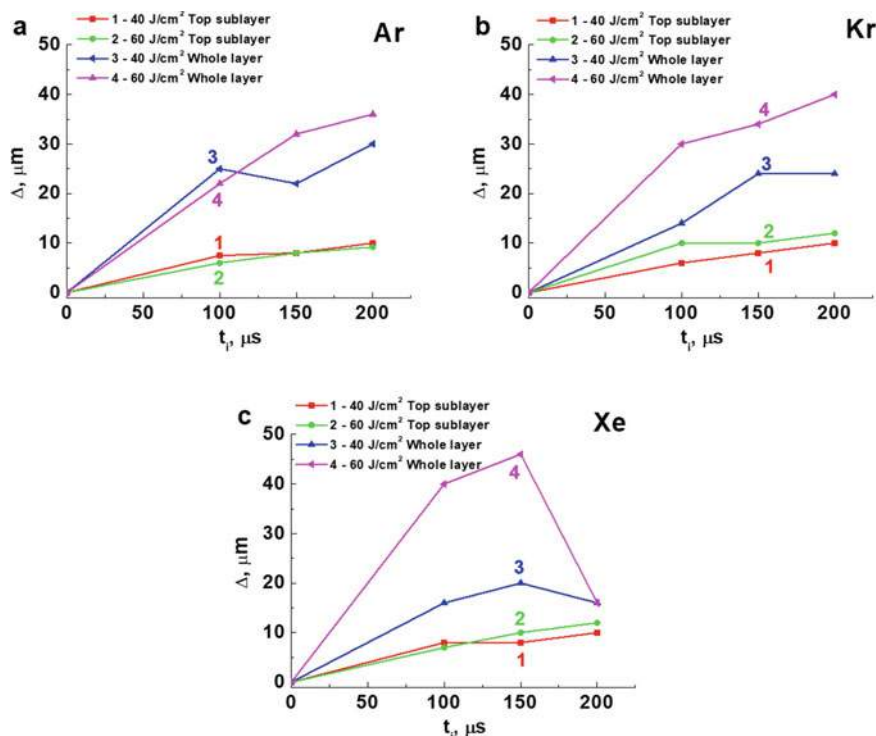


Fig. 18 The thickness of the upper part of the surface layer having the columnar nanostructure (curves 1 and 2) and the whole thickness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer having the modified structure (curves 3 and 4) versus pulse duration (argon, krypton, and xenon plasma, $E_S = 40$ and 60 J/cm^2 , $n = 15$ pulses)

modified surface layer (Figs. 16 and 17). However, the top sublayer thickness was almost independent of both the plasma-forming gas and the electron-beam energy density values.

Figure 19 shows histograms of the size distribution of titanium carbide particles in the top, middle, and bottom sublayers modified in the plasma of argon, krypton, and xenon at an electron-beam energy density of 40 J/cm^2 . The size distribution of titanium carbide particles varied with enhancing distance from the surface to the sample depth. In addition, their average size became larger. It should be noted that this effect was minimal after PEBI in the xenon plasma. The dispersion of the titanium carbide particles almost did not change in the top and middle sublayers, but became a little bit larger in the bottom one. In addition, the greatest thickness of the modified layer was also in the xenon plasma.

Changes in the structure and the phase composition in the surface layer were due to the dissolution of titanium carbide particles in the molten nickel-chrome binder and the formation of a supersaturated with titanium and carbon dissolution at the

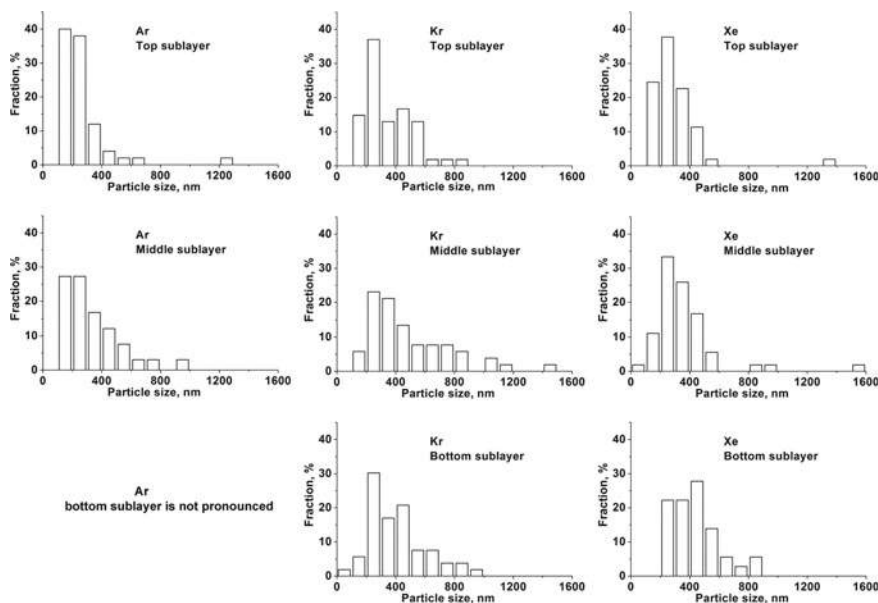


Fig. 19 The histograms of the titanium carbide particle size distribution in the upper, middle and lower parts of the TiC-(Ni-Cr) metal-ceramic alloy modified surface layer after PEBI (argon, krypton, and xenon plasma, $E_S = 40 \text{ J/cm}^2$)

maximum temperature. Also, the secondary titanium carbide nanoparticles precipitated in zones of the primary titanium carbide highest concentration at the interfaces between primary titanium carbide particles and the molten nickel-chrome binder. As a result, a metal-ceramic structure oriented perpendicular to the irradiated surface was formed during solidification under conditions of high temperature gradients. It is shown in Fig. 20 the red line. The primary titanium carbide particles on the surface, where the secondary titanium carbides nanoparticles were formed, are shown by arrows below the red line.

A layer having a metal-ceramic dendrite structure was between the top nanostructured layer and the primary titanium carbide particles partially dissolved in the molten nickel-chrome binder. Elongated titanium carbide particles oriented in different directions were in it (Fig. 21).

The same results were obtained by transmission electron microscopy (TEM). Large titanium carbide particles dispersed to a nanoscale level in the surface layer after PEBI in different plasma-forming gases are shown in Fig. 22. The dispersion of the primary titanium carbide particles occurred by dissolving them in the molten nickel-chrome binder, as well as releasing titanium carbide nanoparticles from a supersaturated with titanium and carbon solid solution in the molten nickel-chrome binder during PEBI in the plasma of argon (a) and krypton (b). The titanium and carbon concentrations in the molten nickel-chrome binder were maximum at the

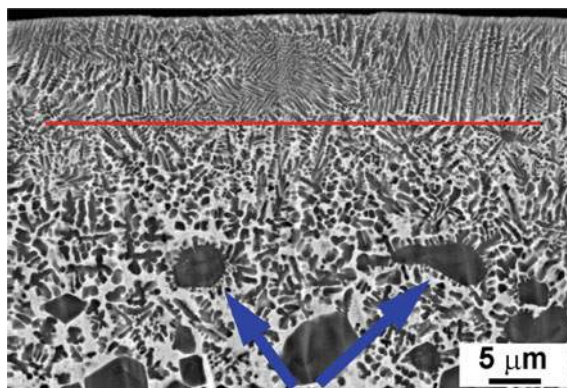


Fig. 20 SEM images of the structure of the TiC-(Ni-Cr) metal-ceramic alloy surface layer after PEBI. Nanosized titanium carbide particles at the interface between the primary titanium carbide particles and the nickel-chrome binder are shown by the arrows

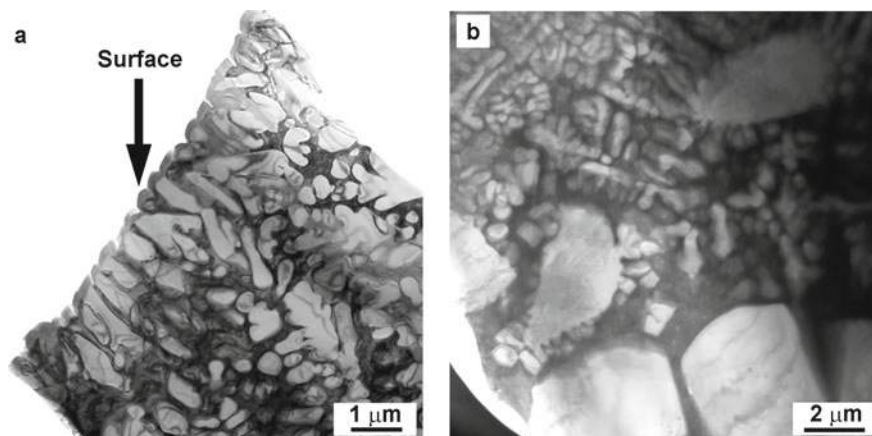


Fig. 21 The dendrite structure of the TiC-(Ni-Cr) metal-ceramic alloy surface layer after PEBI (krypton plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \mu\text{s}$, $n = 15$ pulses)

surfaces of the primary titanium carbide particle. Therefore, the number of titanium carbide nanoparticles was also maximum at the interface between the primary titanium carbide particles and the nickel-chrome binder after solidification (Fig. 22, shown by arrows). As their number increases in the nickel-chrome binder interlayers, the dendritic structure included titanium carbide nanoparticles oriented perpendicular to the irradiated surface that had formed by the temperature field (Fig. 20). The dispersion of the primary titanium carbide particles, as the mentioned mechanism of the titanium carbide nanoparticle formation, apparently occurred by mechanical refining as a result of thermal shock, followed by filling of the discontinuities with the molten nickel-chrome binder upon PEBI in the xenon plasma (Fig. 22c, shown

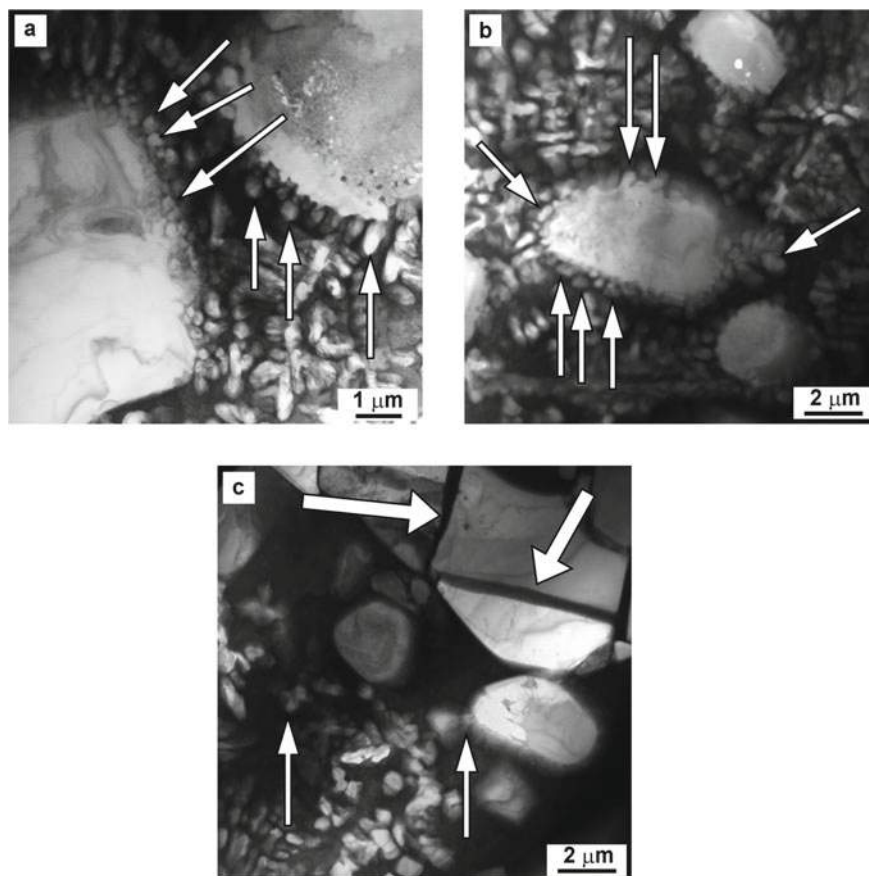


Fig. 22 TEM images of dispersed titanium carbide particles in the initial structure of the TiC-(Ni-Cr) metal-ceramic alloy surface layers after PEBI: **a**—argon plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \mu\text{s}$, $n = 15$ pulses; **b**—krypton plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \mu\text{s}$, $n = 15$ pulses; **c**—xenon plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \mu\text{s}$, $n = 15$ pulses. Nanosized titanium carbide particles formed from a supersaturated solid solution of carbon and titanium on the surface of the primary titanium carbide particles are shown by small arrows. The mechanical failures of the primary carbide particles are shown by the large arrows

by the arrows). Thus, it can be concluded that the heterophase nanostructure was formed in the surface layer of the TiC-(Ni-Cr) metal-ceramic alloy under PEBI in the plasma of light (argon) and heavy (krypton and xenon) inert gases. The modified layer depth increased with the change of the plasma-forming gas from argon to krypton and xenon.

Let us consider X-ray diffraction data. From Table 3, it follows that the phase composition of the material in the initial state (the ceramic to metal component ratio) corresponded to the specified values for the sintered titanium carbide and

Table 3 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer in the initial state

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	37.34	0.35711	49.24	4.685	–
TiC	62.66	0.43122	42.56	2.635	–

nickel-chrome powder mixture. There was no texture in the surface layer of the sample.

PEBI significantly changed the ratio of the ceramic and metal components in the surface layer, as well as formed the (002) texture (Table 4). The presented data were verified by the results of the above numerical estimation of the relative component content in the surface layer. Also, there were a decrease in the nickel-chrome binder content with a corresponding increase in the titanium carbide content, as well as an increase in the lattice parameters of the nickel-chrome binder and titanium carbide. The reason was the mutual solubility of the components when the surface layer was heated under PEBI.

Tables 5 and 6 show diffraction patterns and tables of the structure and phase composition parameters of the surface layers after PEBI in the krypton plasma at electron-beam energy densities of 40 and 60 J/cm². The main features of the structure and the phase composition were a decrease in the titanium carbide content from 90.08 down to 82.51 wt% and corresponding increase in the nickel-chrome binder

Table 4 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (argon plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \mu\text{s}$, $n = 15$ pulses)

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	9.39	0.35900	48.90	3.688	(200)
TiC	90.61	0.43137	59.14	1.181	–

Table 5 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (krypton plasma, $E_S = 40 \text{ J/cm}^2$, $t_i = 200 \mu\text{s}$, $n = 15$ pulses)

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	9.92	0.35772	–	–	(220)
TiC	90.08	0.43037	187.79	2.936	–

Table 6 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (krypton plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 200 \text{ } \mu\text{s}$, $n = 15$ pulses)

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	17.49	0.35778	33.57	3.725	(200)
TiC	82.51	0.43152	20.31	1.953	–

content from 9.92 up to 17.49 wt%. In addition, the titanium carbide lattice parameter increased from 0.43037 up to 0.43152 nm. Also, the size of the coherent scattering regions of the ceramic component decreased from 187.79 down to 20.31 nm. These features became more pronounced after PEBI in the xenon plasma. This can be concluded from the diffraction patterns, as well as the tables of the structure and phase composition parameters presented in Tables 7 and 8 (electron beam energy densities was 40 and 60 J/cm^2 , respectively). An increase in electron-beam energy density from 40 up to 60 J/cm^2 reduced the ceramic content from 93.33 down to 65.00 wt%. This value almost corresponded to the level of its content in the initial state. The lattice parameters increased from 0.35798 up to 0.35920 nm for the nickel-chrome binder and from 0.42844 up to 0.43157 nm for titanium carbide. In addition, the coherent scattering regions for the nickel-chrome binder decreased from 39.01 to 14.36 nm. The lattice microdistortions also decreased from 7.242×10^{-3} down to 1.242×10^{-3} ; texture of the surface layer was the same (002).

Table 7 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (xenon plasma, $E_S = 40 \text{ J/cm}^2$, $t_i = 150 \text{ } \mu\text{s}$, $n = 15$ pulses)

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	6.67	0.35798	39.01	7.280	(200)
TiC	93.33	0.42844	–	–	–

Table 8 The parameters of the structure and the phase composition in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (xenon plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150 \text{ } \mu\text{s}$, $n = 150$ pulses)

Phase type	The relative content (wt%)	The lattice parameter (nm)	The size of the coherent scattering regions (nm)	The lattice micro-distortion, 10^{-3}	Texture
(Ni-Cr)	35.0	0.35920	14.36	1.242	(200)
TiC	65.0	0.43157			

Summarized results of the X-ray phase analysis are presented in Figs. 23 and 24. PEBI in the argon plasma caused a significant decrease in the nickel-chrome binder content in the surface layer, but increased this value by changing the plasma-forming gas to krypton. The nickel-chrome binder content in the surface layer was about the initial state level when the heaviest xenon plasma had been used (Fig. 23a).

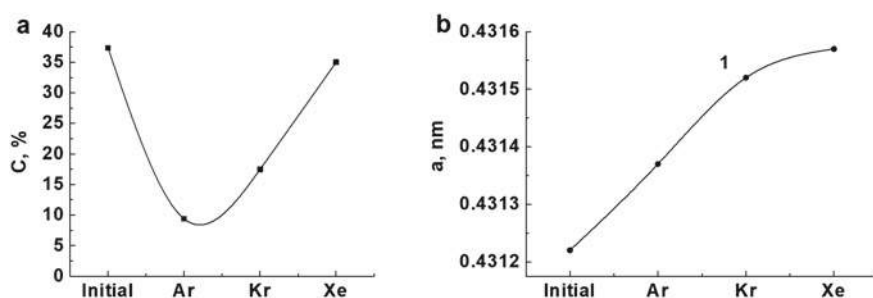


Fig. 23 The relative content of the nickel-chrome binder in the TiC-(Ni-Cr) metal-ceramic alloy surface layer (a) and the crystal lattice parameter of titanium carbide (b) after PEBI ($E_S = 60 \text{ J/cm}^2$, $t_i = 150 \dots 200 \mu\text{s}$, $n = 15$ pulses) versus plasma-forming inert gas: 1—the initial state, 2—argon, 3—krypton, 4—xenon

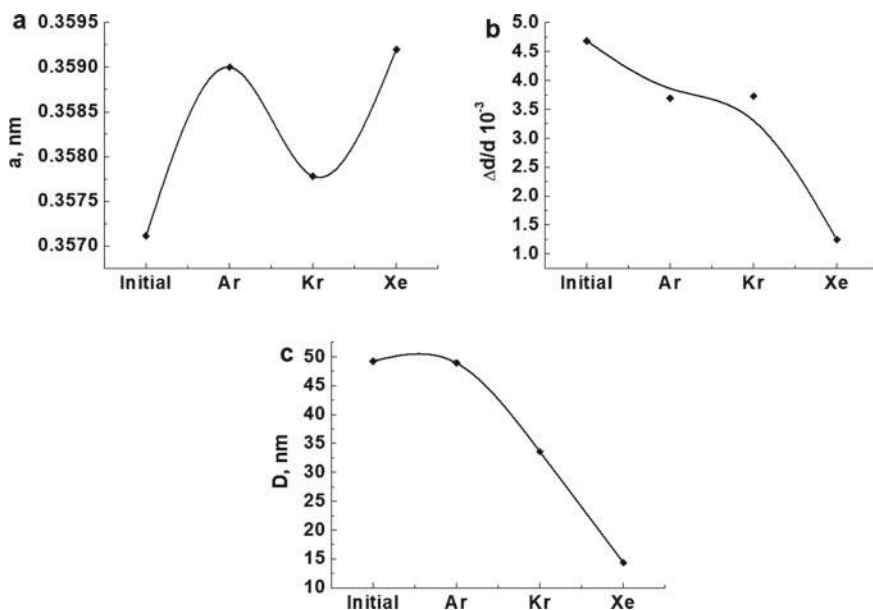


Fig. 24 The lattice parameter (a), the lattice micro-distortion (b), and the size of the coherent scattering regions (c) in the TiC-(Ni-Cr) metal-ceramic alloy surface layer after PEBI ($E_S = 60 \text{ J/cm}^2$, $t_i = 150 \dots 200 \mu\text{s}$, $n = 15$ pulses) versus plasma-forming inert gas: 1—the initial state, 2—argon, 3—krypton, 4—xenon

A feature of the structure and the phase composition of the surface layer after PEBI was an increase in the crystal lattice parameter. This effect intensified with increasing atomic mass of the plasma-forming gas. For example, the titanium carbide crystal lattice parameter reached 0.43157 nm after PEBI in the xenon plasma. This corresponded to the $C/Ti \approx 0.65$ ratio at the maximum titanium carbide hardness.

Figure 24 shows the effect of the plasma-forming gas on the crystal lattice parameters, the lattice microdistortions, and the size of the coherent scattering regions in the nickel-chrome binder. The crystal lattice parameters increased with an increase in atomic mass of the plasma-forming gas as a result of doping of the nickel-chrome binder during its interaction with titanium carbide particles. However, the lattice microdistortions and the sizes of the coherent scattering regions in the nickel-chrome binder decreased with increasing atomic mass of the plasma-forming gas.

3.5 The Effect of the Plasma-Forming Gases on the Nano- and Microhardness, and Wear Resistance of the Modified Surface Layer

Figure 25 shows dependences of nanohardness values in the surface layer after PEBI in the plasma of argon, krypton, and xenon from electron-beam energy density for pulse durations of 100, 150, and 200 μs . Based on their comparison, it can be concluded that the plasma-forming gas had a significant effect on the nanohardness values in the surface layer. They decreased with increasing electron-beam energy density up to 40 J/cm², but then enhanced significantly with rising electron-beam energy density up to 50 J/cm² (regardless of the plasma-forming gas). However, this effect was greater after PEBI in the plasma of krypton or xenon. The maximum nanohardness values were after PEBI in the xenon plasma.

A similar dynamics of the change in the nanohardness values was more concise with a change in pulse duration when electron-beam energy densities were 40 and 60 J/cm² (Fig. 26). Nanohardness values decreased with an increase in pulse duration up to 150 μs (regardless of the plasma-forming gas). Then, they enhanced slightly as pulse duration increased up to 200 μs .

Figure 27 shows dependences of microhardness values in the surface layer after PEBI from pulse duration in the plasma of argon, krypton, and xenon for various values of electron-beam energy density. They have common patterns with the same number of pulses. Initially, the microhardness values enhanced with increasing electron-beam energy density.

However, they change along a curve with a maximum at 150 μs as pulse duration increased, regardless of the plasma-forming gas (Fig. 28). The maximum microhardness values were after PEBI in the plasma of the lightest inert gas (argon). They decreased as the plasma-forming gas changed to krypton and xenon.

The presented data on the effect of atomic mass of the plasma-forming gases on the microhardness values in the surface layer after PEBI enabled to make

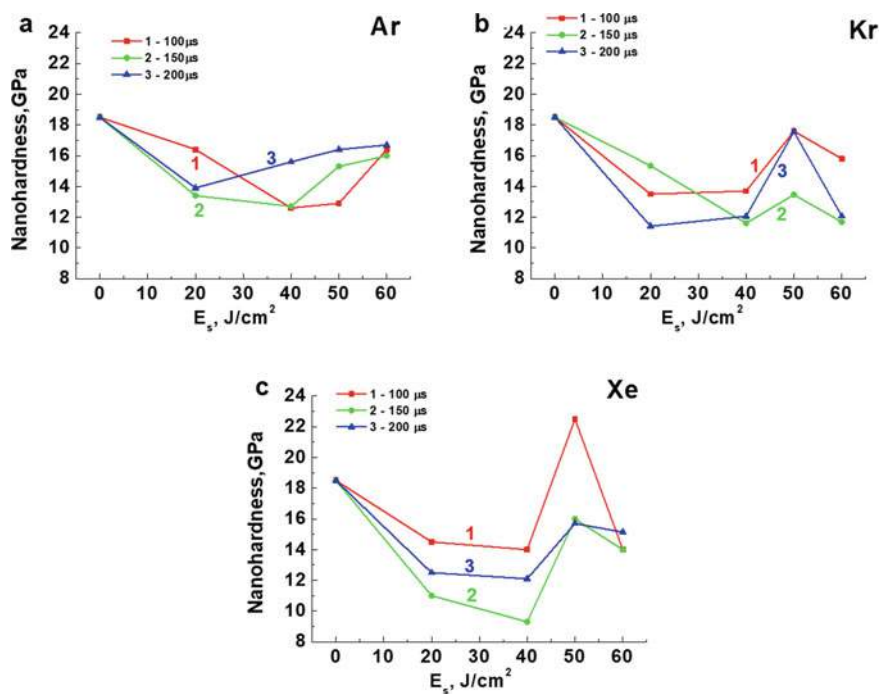


Fig. 25 Nanohardness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer versus electron-beam energy density (argon, krypton, and xenon plasma, $t_i = 100, 150$ and $200 \mu\text{s}$)

conclusions that the best results were obtained at electron-beam energy densities of 40, 50 and 60 J/cm^2 (Fig. 29).

Figure 30 presents dependences of the groove depth from electron-beam energy density after PEBI in the plasma of argon, krypton, and xenon. Modification of the surface layer greatly increased its wear resistance regardless of the plasma-forming gas. However, it increased with enhancing electron-beam energy density. The plasma-forming gas also had a significant effect. The maximum wear resistance values were after PEBI in the xenon plasma at an electron beam energy density of 60 J/cm^2 . It decreased with a change in the plasma-forming gas to krypton and reached the minimum values after PEBI in the argon plasma (Fig. 30).

The presented results were verified by the abrasion test results in accordance with ASTM G65. In the initial period, wear resistance of the samples after PEBI was several times higher than that of the samples in the initial state. The maximum wear resistance was after PEBI in the xenon plasma. However, the modified layer became thinner as the number of revolutions of the abrasive disk increased. As a result, the dendritic structure began to interact with the counterpart and wear resistance of the surface layer decreased to the level of the sample in the initial state.

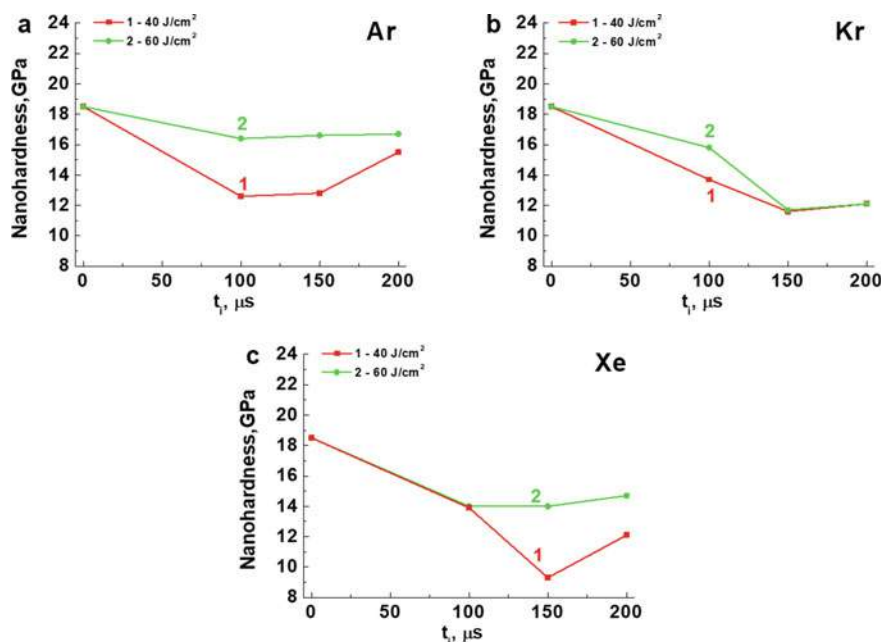


Fig. 26 Nanohardness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer versus pulse duration (argon, krypton, and xenon plasma, $E_S = 40$ and 60 J/cm^2)

Figure 31 shows dependence of the friction coefficient on pulse duration. The main feature of these dependences was a sharp strong decrease in the friction coefficient after PEBI for all studied parameters (regardless of the plasma-forming gas) as compared with the sample in the initial state.

At a pulse duration of $150 \mu\text{s}$, the general pattern of dependences shown in Fig. 32 was the invariance of the friction coefficient with changes in electron-beam energy density and pulse duration in all the plasma-forming gases studied. This means the invariance of the type of the structure and the phase composition of the top part of the surface layer with a change in the PEBI energy parameters.

One of the important parameters of the effect of the surface layer nanostructuring on its tribological properties was the dependence of the friction coefficient on temperature. Figure 33 shows the temperature dependences of the friction coefficient in the initial state and after PEBI by pulses of various durations in the plasma of argon, krypton, and xenon. Modification of the surface layer significantly reduced the friction coefficient over the entire studied temperature range up to 600°C (especially at room temperature). The friction coefficient increased with rising temperature for all samples after PEBI. The friction coefficient of the sample in the initial state changed along a curve with a maximum at 200°C . However, it was larger than that of the samples after PEBI in the entire studied temperature range. The samples after PEBI in the xenon plasma had the smallest values among others.

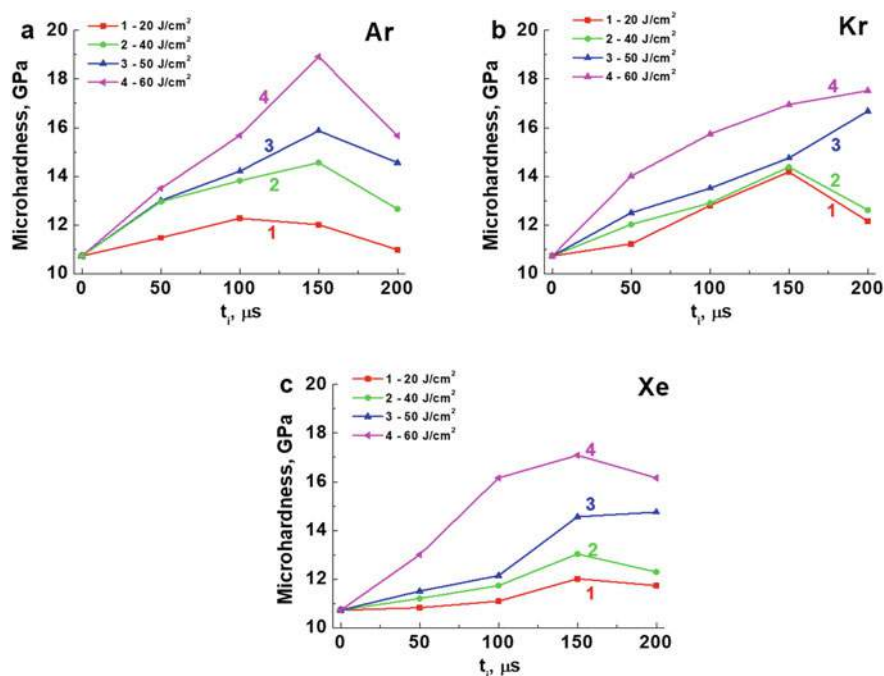


Fig. 27 Microhardness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer versus pulse duration (argon, krypton, and xenon plasma, $E_S = 20, 40, 50$ and 60 J/cm^2 , $n = 15$ pulses)

3.6 The Effect of the Nanostructured Surface Layer on Transverse Bending Strength

It follows from the distribution of the titanium carbide particle sizes, that the top part of the surface layers had the structure included columnar titanium carbide nanoparticles oriented perpendicular to the irradiated surface and the nickel-chrome binder interlayers. It is obvious that an increase in the nanostructuring level in the top part of the surface layer lowered the friction coefficient, and also increased its wear resistance and ductility. As a result, transverse bending strength of the samples increased under loading from the side of the irradiated surface. Figure 34 shows a dependence of transverse bending strength from pulse duration after PEBI. Electron-beam energy density was $40, 50$ and 60 J/cm^2 , the number of pulses was 15, the plasma-forming gases were argon, krypton, and xenon. Modification of the surface layer increased transverse bending strength in all investigated cases. The maximum values (indicated by the ellipses in the graphs) were after PEBI in the xenon plasma with the following combinations of electron-beam energy density and pulse duration: 40 J/cm^2 and $100 \mu s$, 50 J/cm^2 and $200 \mu s$, as well as 60 J/cm^2 and $150 \mu s$.

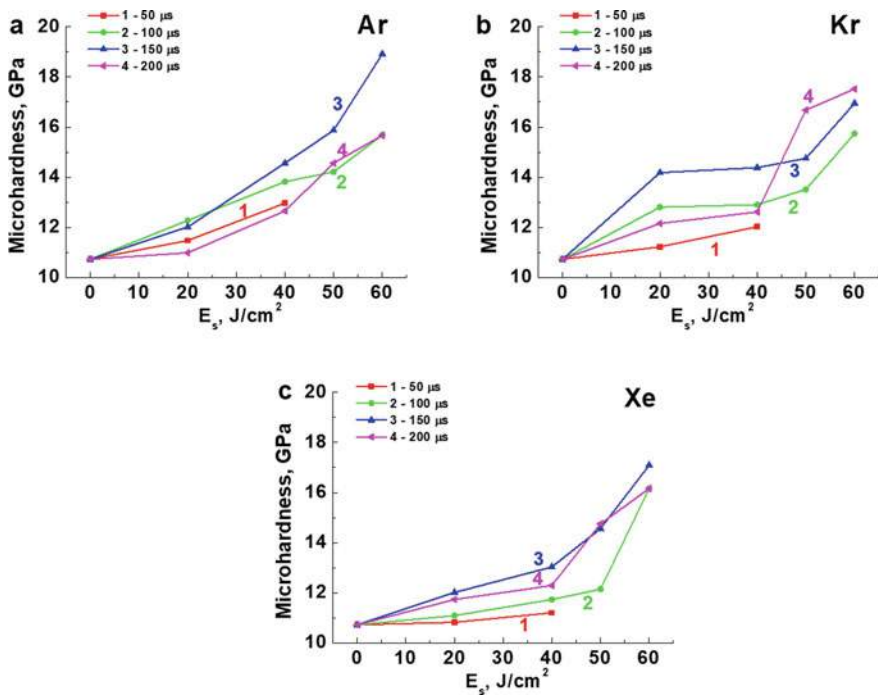


Fig. 28 Microhardness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer versus electron-beam energy density (argon, krypton, and xenon plasma, $t_i = 50, 100, 150$ and 200μ s, $n = 15$ pulses)

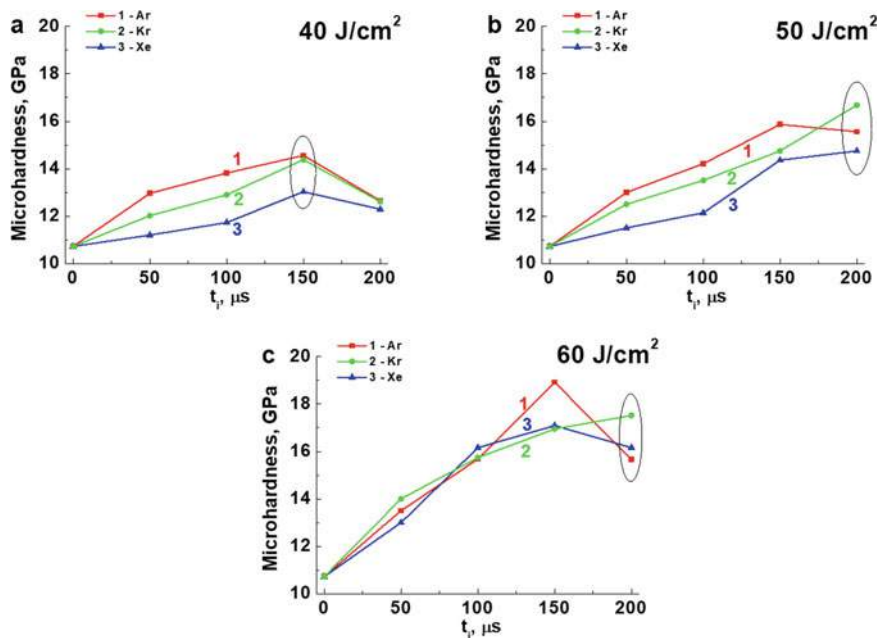


Fig. 29 Microhardness of the TiC-(Ni-Cr) metal-ceramic alloy surface layer versus pulse duration (argon, krypton, and xenon plasma, $E_s = 20, 40$ and 60 J/cm^2 , $n = 15$ pulses). The ovals highlight the areas of greatest values

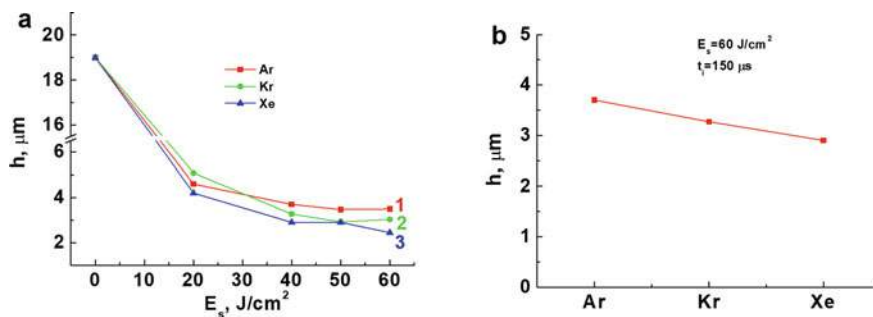


Fig. 30 Depth of the track on the surface of the TiC-(Ni-Cr) metal-ceramic alloy cut with a diamond counterpart after PEBI (argon, krypton, and xenon plasma, $t_i = 150 \mu\text{s}$, $n = 15$ pulses) versus energy density

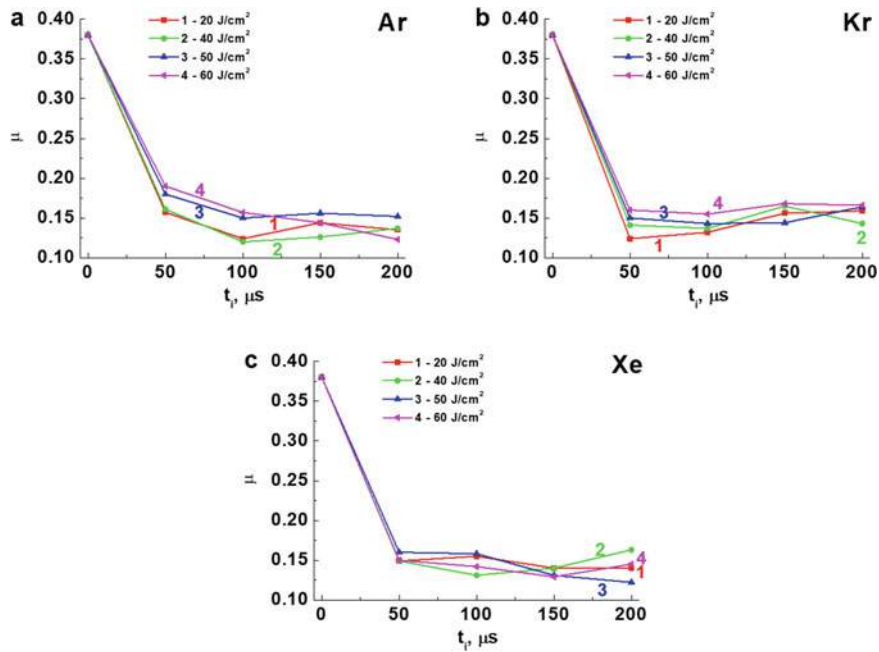


Fig. 31 The friction coefficient of the TiC-(Ni-Cr) metal-ceramic alloy surface after PEBI (argon, krypton, and xenon plasma, $E_S = 20, 40, 50$ and 60 J/cm^2 , $n = 15$ pulses) versus pulse duration

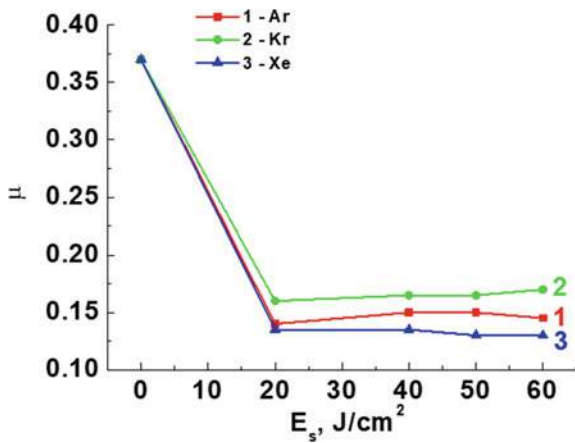


Fig. 32 The friction coefficient of the TiC-(Ni-Cr) metal-ceramic alloy surface after PEBI (argon, krypton, and xenon plasma, $t_i = 150 \mu\text{s}$, $n = 15$ pulses) versus electron-beam energy density

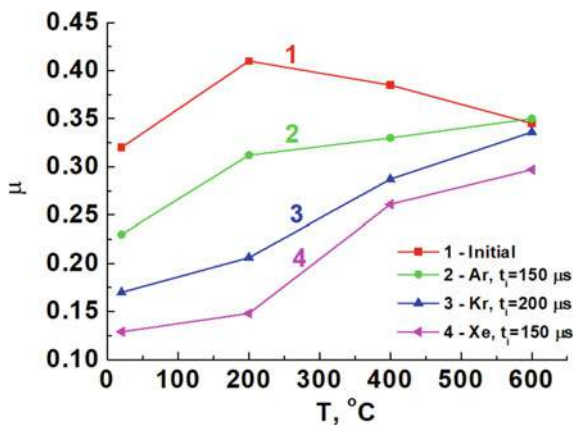


Fig. 33 The friction coefficient of the TiC-(Ni-Cr) metal-ceramic alloy surface after PEBI (argon, krypton, and xenon plasma, $E_S = 60 \text{ J/cm}^2$, $t_i = 150$ and $200 \mu\text{s}$, $n = 15$ pulses) versus temperature

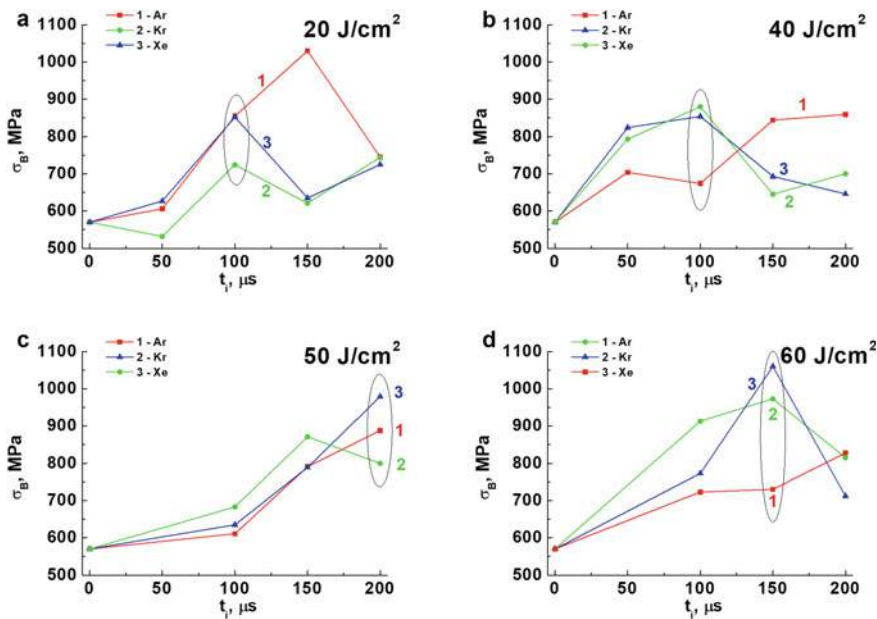


Fig. 34 Bending strength of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI (argon, krypton, and xenon plasma, $E_S = 20, 40, 50$ and 60 J/cm^2 , $n = 15$ pulses, loading to the irradiated side) versus pulse duration

4 Conclusions

The results of the studies of the structure and the phase composition, as well as the physical, mechanical, and tribological properties of the surface layer of the TiC-(Ni-Cr) metal-ceramic alloy after PEBI the inert gas plasma enabled to draw the following conclusions.

1. PEBI in the plasma of the light (argon) and heavy (krypton, xenon) inert gases formed the nanostructured heterophase structure in the surface layer.
2. The modified layer depth increased with the change of the plasma-forming gas from argon to krypton and xenon.
3. Analysis of the results of the electron microscopic studies of the surface layer microstructure showed various mechanisms of the surface layer nanostructuring with an increase in atomic mass of the plasma-forming gas. Formation of titanium carbide nanoparticles directly at the interface of the carbide particles and the molten nickel-chromium binder was supplemented by the mechanical dispersion of titanium carbide particles and filling gaps with the molten nickel-chromium binder.
4. The change in the plasma-forming gas had the significant effect on the nanohardness values in the surface layer after PEBI. They increased greatly with rising atomic mass of the inert gas.
5. The nanohardness values decreased with increasing electron-beam energy density up to 40 J/cm^2 , but then enhanced significantly with rising electron-beam energy density up to 50 J/cm^2 (regardless of the plasma-forming gas). However, this effect was greater after PEBI in the plasma of krypton or xenon. The maximum nanohardness values were after PEBI in the xenon plasma.
6. Typically, the microhardness values changed along the curve with the as pulse duration increased, regardless of the plasma-forming gas. The maximum microhardness values were after PEBI in the plasma of the lightest inert gas (argon). They decreased as the plasma-forming gas changed to krypton and xenon.
7. PEBI in the plasma of the heavy inert gases formed the thicker modified surface layers having lower microhardness values and, accordingly, higher ductility.
8. Modification of the surface layer greatly increased its wear resistance regardless of the plasma-forming gas. However, it increased with enhancing electron-beam energy density. The plasma-forming gas also had a significant effect. The maximum wear resistance values were after PEBI in the xenon plasma at an electron beam energy density of 60 J/cm^2 .
9. The friction coefficient of the nanostructured modified surface layers decreased after PEBI for all studied parameters (regardless of the plasma-forming gas).
10. The general pattern was the invariance of the friction coefficient with changes in electron-beam energy density and pulse duration in all the plasma-forming gases studied. This means the invariance of the type of the structure and the phase composition of the top part of the surface layer with a change in the PEBI energy parameters.

11. Modification of the surface layer by PEBI in the plasma of the inert gases enhanced transverse bending strength of the samples under loading from the side of the irradiated surface.
12. The maximum values of the transverse bending strength were after PEBI in the xenon plasma with the following combinations of electron-beam energy density and pulse duration: 40 J/cm² and 100 μ s, 50 J/cm² and 200 μ s, as well as 60 J/cm² and 150 μ s.
13. The maximum values of the transverse bending strength were after PEBI in the xenon plasma when electron-beam energy density was 60 J/cm². Transverse bending strength increased from 570 MPa in the initial state up to 1061 after PEBI (almost doubled).

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Adhesion of a Thin Soft Matter Layer: The Role of Surface Tension



Valentin L. Popov

Abstract We consider an adhesive contact between a thin soft layer on a rigid substrate and a rigid cylindrical indenter (“line contact”) taking the surface tension of the layer into account. First, it is shown that the boundary condition for the surface outside the contact area is given by the constant contact angle—as in the case of fluids in contact with solid surfaces. In the approximation of thin layer and under usual assumptions of small indentation and small inclination angles of the surface, the problem is solved analytically. In the case of a non-adhesive contact, surface tension makes the contact stiffer (at the given indentation depth, the contact half-width becomes smaller and the indentation force larger). In the case of adhesive contact, the influence of surface tension seems to be more complicated: For a flat-ended punch, it increases with increasing the surface tension, while for a wedge, it decreases. Thus, the influence of the surface tension on the adhesion force seems to be dependent on the particular geometry of the contacting bodies.

Keywords Adhesion · Capillarity · Surface tension · Winkler foundation · Contact angle

1 Introduction

Classical contact mechanics as represented by the works of Hertz [1] or Bussinesq [2], see also [3], neglects the surface tension of the contacting solids. In reality, all three surfaces of bodies in contact (Fig. 1a) can be characterized by their specific surface energies γ_1 , γ_2 and γ_{12} . Depending on their values, one can distinguish several cases. If the specific surface energy of the surface of elastic body outside the contact area can be neglected, then we have an adhesive contact with specific work of separation $w = \gamma_2 - \gamma_{12}$. This case was first considered in the classic work by Johnson, Kendall and Roberts [4]. If the surface energy of the elastic body outside the contact area is finite, $\gamma_1 \neq 0$, but the work of adhesion, which in the general case is equal to

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$$w = \gamma_1 + \gamma_2 - \gamma_{12} \quad (1)$$

is zero, then we have a non-adhesive contact with surface tension. One can consider such system as an elastic body coated with a stressed membrane. The corresponding theory was first developed in [5]. The general case is when both work of adhesion and surface tension of the “free surface” are finite. This leads to a general adhesive contact with surface tension. The latter attracted much interest in the last two decades in the context of indentation of soft matter (e.g. gels or biological tissues) [6–8]. Let us also note that another contact problem with adhesion and surface tension represents a contact of an elastic solid with a *fluid* [9]. This class differs significantly from contact of elastic bodies with surface tension and will not be discussed here.

2 Model

In the present paper, we consider an adhesive contact between a thin soft layer on a rigid substrate and a rigid cylindrical indenter (“line contact”) taking the surface tension of the layer into account, Fig. 1a (left). Without consideration of the surface tension, this problem has been solved in [10]. Here we extend the study carried out in [10] to include the effect of surface tension of contacting bodies. This contact problem can be treated asymptotically exactly, but only under strong assumptions. In particular, we assume that the following conditions are fulfilled: $d \ll h$, $h \ll a$, where d is the indentation depth, h the thickness of the layer, and a the half-width of the contact (definitions see in Fig. 1a (left)). Additionally, it is assumed that the slope of the profile of contacting bodies and of the free surface outside the contact

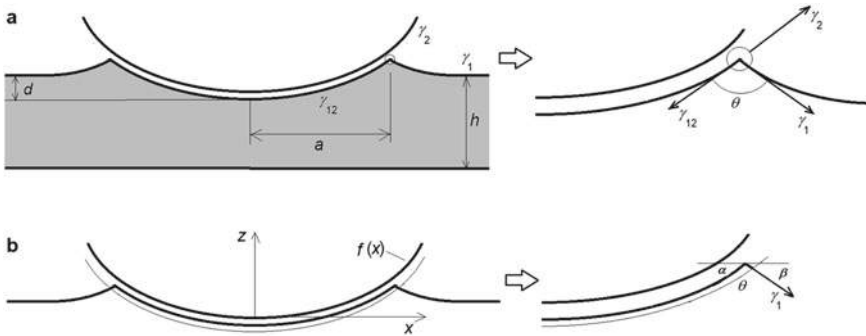


Fig. 1 **a** left: principal sketch of the system consisting of a rigid indenter in contact with a thin elastic layer (grey) having the initial thickness h . **a** right: Detailed view of the boundary of the contact with three surface forces corresponding to the three surfaces meeting at the boundary. **b** left: Definitions of coordinates and profile of the rigid indenter as well as free body diagram (of the system over the thin gray line). **b** right: Detailed picture of a part of the free body diagram showing the surfaces forces acting on the contact boundary

is much smaller than unity. Further conditions, if necessary, will be specified later in this paper.

Under the above conditions, the elastic layer is deformed uniaxially, independently in each point and the layer can be considered as a two-dimensional elastic foundation with effective modulus [11]

$$\tilde{E} = \frac{E(1 - \nu)}{(1 + \nu)(1 - 2\nu)}, \quad (2)$$

where E is elastic modulus and ν Poisson number. Due to local uniaxial deformation, the layer can be considered as a two-dimensional elastic foundation composed of independent springs placed with separation Δx and Δy correspondingly, while each spring has the stiffness

$$\Delta k = \tilde{E} \frac{A}{h}, \quad (3)$$

with $A = \Delta x \Delta y$. When a rigid profile $f(x)$ (Fig. 1b (left)) is indented into this elastic foundation by a depth d , then the vertical displacements of the springs in contact are equal to

$$u_z(x) = d - f(x), \quad |x| \leq a \quad (4)$$

(Note that while the axis z for the definition of the profile shape is directed upwards, the positive direction of the displacement $u_z(x)$ is accepted to be downwards).

The shape of the surface outside the contact area, is governed by the equation

$$\gamma_1 \frac{\partial^2 u_z(x)}{\partial x^2} = \frac{\tilde{E}}{h} u_z(x), \quad (5)$$

which simply equates the elastic stress to the stress produced by the tensioned surface (surface tension γ_1 multiplied with the surface curvature $\partial^2 u_z(x)/\partial x^2$). Solution of Eq. (5) reads

$$u_z(x) = -C \exp(-x/l), \quad (6)$$

where

$$l = \sqrt{\frac{\gamma_1 h}{\tilde{E}}}. \quad (7)$$

This length plays the role of the “elastocapillary length” in the present problem.

3 Boundary Condition at the Contact Boundary

Equation (5) must be completed through boundary conditions at the boundary of the contact area. To derive this boundary condition, consider a small part of the boundary encircled in Fig. 1a (right) by a gray circle. The sum of all forces acting on the boundary line parallel to the surface of the rigid body, should vanish, if the boundary friction is neglected:

$$\gamma_{12} + \gamma_1 \cos \theta - \gamma_2 = 0. \quad (8)$$

The elastic force can be neglected in this equation as it vanishes if the size of the circle tends towards zero (while the surface tensions remain constant). This equation is the same as in the case of a contact of a liquid with a solid. Karpitschka et al. come to the same conclusion by performing minimization of the complete energy functional [12].

4 The Force Acting on the Rigid Indenter

As we consider the line contact (no dependency on the coordinate perpendicular to the plane (x, z) (not shown in Fig. 1)), it is convenient to use instead of the normal force the normal force per length, P_N and also normalize all other forces per unit length. The elastic force per unit length acting on the rigid indenter from the elastic layer is simply:

$$P_{el} = \frac{\tilde{E}}{h} \int_{-a}^a u_z(x) dx = \frac{\tilde{E}}{h} \int_{-a}^a (d - f(x)) dx. \quad (9)$$

Apart from this elastic force, there is an additional force acting on the indenter by the surface of the elastic body outside the contact area, which is equal to (see Fig. 1b (right))

$$P_{\text{surf}} = -2\gamma_1 \sin \beta. \quad (10)$$

Under assumption that all slopes in the considered system are small, the total force acting on the rigid indenter from the elastic body is equal to

$$P_N = 2 \frac{\tilde{E}}{h} \int_0^a (d - f(x)) dx - 2\gamma_1 \left. \frac{\partial u_z(x)}{\partial x} \right|_{x=a+0}. \quad (11)$$

In this equation, the half-width of the contact, a , is still not defined.

5 Contact Half-Width

The boundary condition for the surface shape can be written as

$$\left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0} + \left. \frac{\partial u_z(x)}{\partial x} \right|_{x=a+0} + \theta = \pi. \quad (12)$$

Note that we assume that all slopes are small, so that the contact angle should be almost equal to π . Smaller contact angles can be realized physically but they cannot be treated in the approximation of small slopes, which is used in the present model (see, however, the next section for a more detailed discussion of the area of applicability).

At the boundary of the contact area two equations have to be fulfilled:

$$d - f(a) = -C \exp(-a/l), \quad (13)$$

and

$$\left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0} + \frac{C}{l} \exp(-a/l) + \theta = \pi. \quad (14)$$

Substituting (13) as well as the solution of Eq. (8),

$$\pi - \theta \approx \sqrt{\frac{2w}{\gamma_1}}, \quad (15)$$

into (14) gives

$$f(a) = d + \sqrt{\frac{2wh}{\tilde{E}}} - \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0}. \quad (16)$$

Here

$$w = \gamma_1 + \gamma_2 - \gamma_{12} \quad (17)$$

is the work of adhesion.

In the limit $\gamma_1 = 0$ (vanishing surface tension), (16) reduces to

$$f(a) = d + \sqrt{\frac{2wh}{\tilde{E}}}, \quad (18)$$

which, according to [10], is the correct result for an arbitrary profile if the surface tension is neglected.

6 Area of Applicability of Eq. (16)

Note that all above considerations are valid under the assumptions listed in Sect. 2. In particular, the thin layer approximation can only be used if all slopes are small. For our problem, this implies that the angle $\pi - \theta$ should also be small. From Eq. (15) it then follows that the current approximation is strictly valid only if $2w \ll \gamma_1$. This means that the limit $\gamma_1 \rightarrow 0$ is not covered by the present theory. However, Eq. (18), obtained in the limit $\gamma_1 = 0$, reproduces the exact solution of the corresponding problem without surface tension. This suggests that Eq. (16) can be used as an approximate solution in the whole range of values of $0 \leq \gamma_1 \leq \infty$.

7 Case Studies

Case study 1: rigid plane In this case, $f(x) = 0$ and Eq. (16) takes the form

$$0 = d + \sqrt{\frac{2wh}{\tilde{E}}}. \quad (19)$$

For negative indentation depth (which correspond to the adhesion case), this equation is fulfilled for one single value of the distance between the rigid plane and the elastic layer:

$$|d_c| = -d_c = \sqrt{\frac{2wh}{\tilde{E}}}. \quad (20)$$

If the distance becomes larger, the contact shrinks and disappears; if it becomes smaller, then it spreads to infinity. At exactly the critical value, the contact is in an indefinite equilibrium at any contact size. These properties are the same as in the case of adhesive contact without tension.

Case study 2: flat punch with half-width a As is clear from the Case study 1, a flat ended punch will detach at once at the critical distance, given by Eq. (20). The surface shape outside the contact is given by

$$u_z(x) = -\sqrt{\frac{2wh}{\tilde{E}}} \exp\left(\frac{-x+a}{l}\right). \quad (21)$$

Equation (11) for the normal force now gives

$$P_N = -2a\sqrt{\frac{2w\tilde{E}}{h}} - 2\sqrt{2w\gamma_1} = -2^{3/2}w^{1/2}\left(a\sqrt{\frac{\tilde{E}}{h}} + \sqrt{\gamma_1}\right). \quad (22)$$

The force of adhesion is minus the normal force acting on the rigid indenter:

$$P_A = 2^{3/2}w^{1/2}\left(a\sqrt{\frac{\tilde{E}}{h}} + \sqrt{\gamma_1}\right). \quad (23)$$

We see that, at the given work of separation w , the surface tension leads to an increase of the force of adhesion. The contact configuration is illustrated in Fig. 2.

Case study 3: wedge shape If the shape of the rigid indenter is given by $f(x) = |x| \tan \delta$ (Fig. 3a), then Eq. (16) takes the form

$$a \tan \delta = d + \sqrt{\frac{2wh}{\tilde{E}}} - \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \tan \delta \quad (24)$$

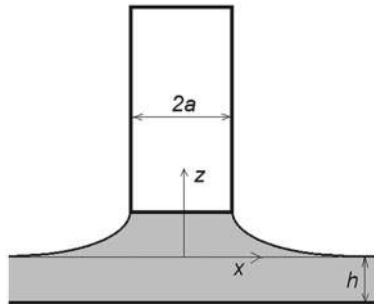


Fig. 2 Adhesive contact with surface tension of a flat rigid indenter with a thin elastic layer

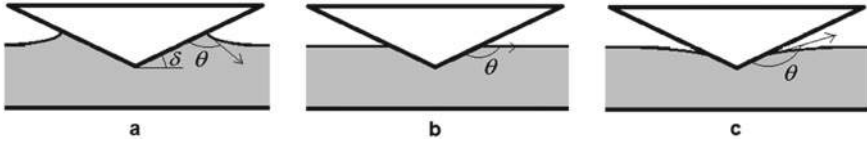


Fig. 3 Contact of a wedge shaped indenter for three different contact angles. Case (b) correspond to the vanishing adhesion force

and the half-width of the contact area is given by

$$a = (\tan \delta)^{-1} \left(d + \sqrt{\frac{2wh}{\tilde{E}}} \right) - \sqrt{\frac{\gamma_1 h}{\tilde{E}}}. \quad (25)$$

The shape of the surface outside the contact can be written as

$$u_z(x) = \left(-\sqrt{\frac{2wh}{\tilde{E}}} + \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \tan \delta \right) \exp\left(\frac{-x + a}{l}\right). \quad (26)$$

For the normal force, Eq. (11), we get:

$$P_N = 2 \frac{\tilde{E}}{h} \left(ad - \frac{a^2}{2} \tan \delta \right) + 2 \left(-\sqrt{2w\gamma_1} + \gamma_1 \cdot \tan \delta \right). \quad (27)$$

Solving Eq. (24) with respect to d ,

$$d = a \tan \delta - \sqrt{\frac{2wh}{\tilde{E}}} + \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \tan \delta \quad (28)$$

and inserting this result into (27) gives the force per length as a function of the contact half-width a :

$$P_N = 2 \frac{\tilde{E}}{h} \left(\frac{a^2}{2} \tan \delta - a \left(\sqrt{\frac{2wh}{\tilde{E}}} - \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \tan \delta \right) \right) + 2 \left(-\sqrt{2w\gamma_1} + \gamma_1 \cdot \tan \delta \right). \quad (29)$$

Minimizing with respect to a , gives the adhesion force

$$F_A = |P_{N,\min}| = \frac{2w}{\tan \delta} - \gamma_1 \tan \delta. \quad (30)$$

Surface tension leads to a decrease of the force of adhesion. The adhesion force vanishes when

$$\tan \delta = \sqrt{\frac{2w}{\gamma_1}}, \quad (31)$$

or, under consideration of (15) and assuming $\tan \delta \approx \delta$,

$$\delta + \theta = \pi. \quad (32)$$

This equation has a very simple physical interpretation: The adhesion force disappears if the slope of the indenter and the contact angle allow a horizontal non-deformed surface outside the contact area, as illustrated in Fig. 3b. Under condition (31) the whole dependence of the force on the contact half-width, Eq. (29), is reduced to that for the non-adhesive contact without surface tension:

$$P_N = \frac{\tilde{E}}{h} a^2 \tan \delta. \quad (33)$$

This is because, in this case, the surface tension force is directed horizontally and does not contribute to the normal force.

Case study 4: parabolic shape Let us consider the special case of a parabolic profile

$$f(x) = \frac{x^2}{2R}. \quad (34)$$

Equation (16) now takes the form

$$\frac{a^2}{2R} = d + \sqrt{\frac{2wh}{\tilde{E}}} - \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \frac{a}{R} \quad (35)$$

Its solution with respect to a reads

$$a = -\sqrt{\frac{\gamma_1 h}{\tilde{E}}} + \sqrt{\frac{\gamma_1 h}{\tilde{E}} + 2R \left(d + \sqrt{\frac{2wh}{\tilde{E}}} \right)} \quad (36)$$

In the case of vanishing surface tension, $\gamma_1 = 0$, Eq. (36) reduces to $a = \sqrt{2R \left(d + \sqrt{\frac{2wh}{\tilde{E}}} \right)}$ meaning that the contact boundary is defined by cutting the profile at the height $d + \sqrt{\frac{2wh}{\tilde{E}}}$ which coincides with the result of paper [10] for the corresponding problem with vanishing surface tension.

8 Non-adhesive Contact

Let us consider the limiting case of non-adhesive contact with tension separately. Under “non-adhesive” contact we will understand the contact of surfaces with vanishing work of separation, $w = 0$. From the definition (1), it follows that in this case

$$\gamma_1 = \gamma_{12} - \gamma_2. \quad (37)$$

From (8), it follows that

$$\cos \theta = \frac{\gamma_2 - \gamma_{12}}{\gamma_1} = -1 \quad (38)$$

and $\theta = \pi$. This means that the slope is continuous at the boundary of the contact. Equation (14) now takes the form

$$\left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0} + \frac{C}{l} \exp(-a/l) = 0. \quad (39)$$

Taking (13) into account, we come to the equation

$$d - f(a) = l \cdot \left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0} = \sqrt{\frac{\gamma_1 h}{\tilde{E}}} \cdot \left. \frac{\partial f(x)}{\partial x} \right|_{x=a-0}. \quad (40)$$

Consider as an example a contact of a parabolic indenter $f(x) = x^2/(2R)$. Equation (40) takes the form

$$a^2 + 2al - 2Rd = 0 \quad (41)$$

For the contact radius we thus get

$$a = -l + \sqrt{2Rd + l^2} \quad (42)$$

This means that the surface tension leads to a decrease of the contact width compared with the non-adhesive contact without surface tension.

For the total normal force we get according to (11)

$$P_N = \frac{2\tilde{E}}{3Rh} \left[(2Rd + l^2)^{3/2} - l^3 \right] = \frac{2\tilde{E}}{3Rh} \left[\left(2Rd + \frac{\gamma_1 h}{\tilde{E}} \right)^{3/2} - \left(\frac{\gamma_1 h}{\tilde{E}} \right)^{3/2} \right] \quad (43)$$

For given d , the normal force with surface tension is larger than that without surface tension.

9 Conclusion

In the present paper, we considered a general adhesive contact of a thin elastic body with a rigid indenter. An important conclusion is that at the boundary of the contact area, the surface of the elastic layer meets the surface of the rigid indenter under a fixed contact angle, which is determined uniquely by the specific surface energies of the rigid body, the elastic body and the interface. The solution obtained for very small surface tension (compared with the work of adhesion) seems to provide a good approximation for arbitrary values of the surface tension. In the case of a non-adhesive contact, surface tension makes the contact stiffer (at the given indentation depth, the contact half-width becomes smaller and the indentation force larger). In the case of adhesive contact, the influence of surface tension seems to be more complicated: For a flat-ended punch, it increases with increasing the surface tension, while for a wedge, it decreases. Thus, the influence of the surface tension on the adhesion force seems to be dependent on the particular geometry of the contacting bodies.

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Adhesion Hysteresis Due to Chemical Heterogeneity



Valentin L. Popov

Abstract According the JKR theory of adhesive contact, changes of the contact configuration after formation of the adhesive neck and before detaching are completely reversible. This means, that after formation of the initial contact, the force-distance dependencies should coincide, independently of the direction of the process (indentation or pull-off). In the majority of real systems, this invariance is not observed. The reasons for this may be either plastic deformation in the contacting bodies or surface roughness. One further mechanism of irreversibility (and corresponding energy dissipation) may be chemical heterogeneity of the contact interface leading to the spatial dependence of the specific work of adhesion. In the present paper, this “chemical” mechanism is analyzed on a simple example of an axisymmetric contact (with axisymmetric heterogeneity). It is shown that in the asymptotic case of a “microscopic heterogeneity”, the system follows, during both indentation and pull-off, JKR curves, however, corresponding to different specific surface energies. After the turning point of the movement, the contact area first does not change and the transition from one JKR curve to the other occurs via a linear dependency of the force on indentation depth. The macroscopic behavior is not sensitive to the absolute and relative widths of the regions with different surface energy but depends mainly on the values of the specific surface energy.

Keywords Adhesion · Hysteresis · Energy dissipation · JKR theory · MDR · Specific surface energy · Heterogeneity

1 Introduction

Johnson, Kendall and Roberts published 1971 their famous work on adhesive contact of elastic parabolic bodies [1]. Contrary to the non-adhesive contact, an adhesive contact shows a hysteresis: the dependencies of force on approach depend on whether the bodies are brought into contact or pulled off. The area enclosed in the hysteresis

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loop is the energy, which is irreversible dissipated during one complete “cycle” of an adhesive contact. According to the JKR theory, after the first contact, an adhesive neck of finite radius appears. If we now would try to pull off the bodies, they remain in contact even for negative values of the indentation depth up to the point of instability where the contact is lost at once. The mechanical energy is irreversibly lost only in such points of instabilities [2]. Both before and after the instability, the processes of approach and detachment are reversible, which is obvious if we remember that the JKR theory is based on the principle of virtual work, which assumes absence of static frictional forces [3]. However, experiments show that adhesive contacts show often pronounced hysteresis even after the formation of initial contact [4–6]. Some authors attribute this to plastic deformation [5]. However, it was shown in [6, 7] that such hysteresis could be seen also in pure elastic contacts between rough bodies. The mechanism of this hysteresis is very simple: The energy dissipation occurs in each act of instable movement of the contact boundary. If the contact area has a complicated shape, the movement of the boundary can proceed in a series of jumps [8] leading to energy dissipation, which means that the boundary feels a dissipative force (see also the supplementary video [9]). Another mechanism of the adhesive hysteresis in the already formed contact state could be the chemical heterogeneity of the contact interface. In the present paper, we analyze this mechanics on a simple example when both the shape of contacting bodies and the chemical heterogeneity have axial symmetry.

2 Problem Statement and Model Description

Consider an adhesive contact between an axisymmetrical rigid body $z = f(r)$, where z is the coordinate in the normal to interface direction and r polar radius in the contact plane, and an elastic half space. It is assumed that the specific work of adhesion assumes two constant values γ_1 and γ_2 in the alternating rings having the widths h_1 and h_2 (Fig. 1). In the framework of the Method of Dimensionality Reduction (MDR) [10, 11], it is possible to map a three-dimensional axisymmetric contact problem to a contact of the modified plane shape

$$g(x) = |x| \int_0^{|x|} \frac{f'(r)}{\sqrt{x^2 - r^2}} dr. \quad (1)$$

and a one-dimensional elastic foundations consisting of independent springs (Fig. 2) having the stiffness

$$\Delta k = E^* \Delta x. \quad (2)$$

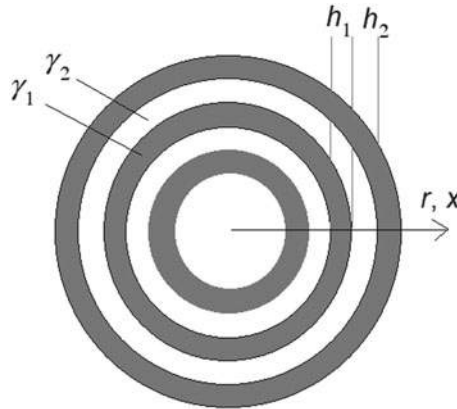


Fig. 1 Schematic representation of the chemical heterogeneity in the considered system. The specific work of adhesion take two constant values γ_1 and γ_2 in the alternating rings having the widths h_1 and h_2

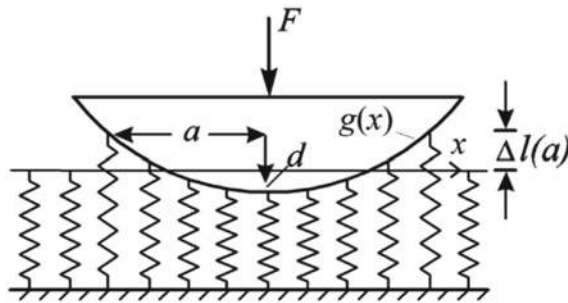


Fig. 2 Scheme of the MDR-representation of an adhesive contact. The equivalent profile $g(x)$ given by (1) is brought into contact with elastic foundation defined by Eq. (2). The contact radius is defined by the condition that the elongation of the springs at the boundary of the contact is given by the rule of Heß, Eq. (5). The normal force, contact radius and the indentation depth in this MDR-model are the same as in the initial three-dimensional contact problem

Here Δx is the space between two adjacent springs, and

$$E^* = E/(1 - \nu^2), \tag{3}$$

with E the Young modulus and ν the Poisson number of the elastic half-space.

In the MDR, it can be shown [10, 11] that the indentation depth, the contact radius and the normal force calculated as a sum of forces of all springs in contact:

$$F(a) = 2E^* \int_0^a (d - g(\tilde{a})) d\tilde{a} \tag{4}$$

coincide with their values in the original three-dimensional problem. The radius of the adhesive contact is determined from the requirement of the minimum of the total energy of the system. This means that if detachment of two springs on both sides of the contact is leading to a decrease of the total energy (elastic energy plus surface energy) then it will detach. On the other hand, if the formation of contact for the springs adjacent to those at the edge of the contact, leads to a decrease of energy, the contact will spread further. Detachment of two springs leads to a decrease of elastic energy by $E^* \cdot \Delta x \cdot \Delta l^2$, where Δl is the elongation which a spring has in the attached state (Fig. 2). When it detaches, a free surface having the area $2\pi a \Delta x$ is formed, which increases the energy by the work of separation $2\pi a \Delta x \gamma$. The boundary is in equilibrium if these two energies are equal and thus

$$\Delta l = \sqrt{\frac{2\pi a \gamma}{E^*}}. \quad (5)$$

This equation, which is equivalent to the Griffith criterion for crack equilibrium [12], was first found first found by Heß [13] and is known as *rule of Heß* [14]. Using the relation $u(x) = d - g(x)$, where $u(x)$ is the vertical displacement at the position x , we can rewrite (5) in the form

$$d = g(a) - \sqrt{\frac{2\pi a \gamma}{E^*}}. \quad (6)$$

This equation connects the indentation depth with the equilibrium contact radius, a . In the following, for simplicity, we will assume that the contact is realized by a very stiff system, which means that the indentation d can be considered as controlling parameter.

3 Attachment and Detachment of a Chemically Heterogeneous Body

Consider the system with specific surface energy depending on the position as shown in Fig. 1. Assume $\gamma_1 < \gamma_2$. During the indentation, there are three repeating stages in the movement (Fig. 3).

(1) If at some moment of time the contact radius a_1 coincides with the inner edge of the ring having the surface energy γ_1 , then at this moment

$$d_1 = g(a_1) - \sqrt{\frac{2\pi a_1 \gamma_1}{E^*}}. \quad (7)$$

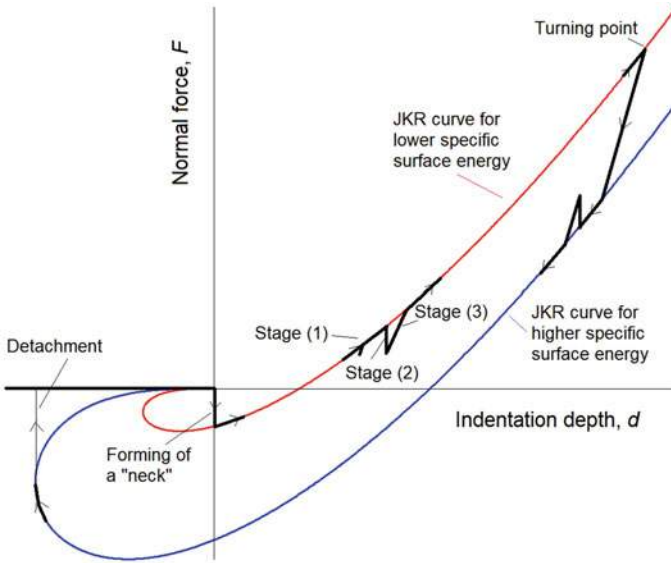


Fig. 3 Processes of approach, formation of contact and pull-off for a heterogeneous contact. During the Stage (1), the boundary moves reversibly along the ring with lower specific surface energy. During the Stage (2) it jumps over the ring with higher specific surface energy. During the Stage (3) it “sticks” in this position until the force reaches the JKR curve. This movement occurs at a constant contact radius and is thus linear. After that, this quasi-periodic process is repeated (the repetitions are not shown in the Figure). If the direction of movement is changed to the opposite (“turning point”), the contact radius first remains constant causing a linear dependency of the normal force on approach. After the force have reached the JKR curve corresponding to the higher specific surface energy, the process consisting of reversible propagation inside the rings with high specific surface energy, jumps over the rings with low specific surface energy and linear returns to the JKR curve

If the indentation depth increases, the contact radius will also increase (exactly accordingly to the corresponding JKR curve with surface energy γ_1 , Fig. 3, Stage (1)) unless it reaches the outer edge of the ring having the surface energy γ_1 . At this moment

$$d_2 = g(a_2) - \sqrt{\frac{2\pi a_2 \gamma_1}{E^*}}, \quad a_2 = a_1 + h_1. \quad (8)$$

(2) Further increasing of indentation depth leads to a jump-like increase of the surface energy. Therefore, the contact boundary will jump over the whole width of the ring with higher surface energy (at the given indentation depth (8)) and stop at the edge of the ring having lower surface energy. At this point, the configuration is given by the pair

$$(d_2, a_2 + h_2). \quad (9)$$

This jump in the contact area will lead to a (negative) jump in the force (see Fig. 3, Stage (2)).

(3) During further indentation, the contact radius will remain constant and the force will therefore increase linearly with indentation depth until it reaches again the JKR curve.

After that, we are again in the repetition of the Stage (1), and the movement occurs along the JKR curve to the next jump, and so on.

We see that in the phase of indentation the system follows the JKR curve corresponding to the *lower surface energy*, with periodic negative jumps and linear returns to the JKR curve. The maximum amplitude of a jump corresponds to the “distance” between the JKR curves for γ_1 and γ_2 . In the following, we assume that the amplitude of jumps is small compared to this “distance”. Under this assumption, the indentation occurs practically along the JKR curve for smaller surface energy with small variations.

If at some point the indentation stops and reversed movement starts, then the system first remains stuck in this point. This is because in spreading, the contact area is pinned by the areas with lower specific surface energy while in detaching, it is pinned by the areas with higher specific surface energy. Therefore, the transition from indentation to pulling off first leads to the “switching of the criterion for propagation” which leads to pinning the boundary to the position at the beginning of reverse motion. As the contact area remains constant, the force-distance dependency is in this stage linear until the force reaches the JKR curve corresponding to the higher specific surface energy. In the following, it moves along the JKR curve corresponding to the higher specific surface energy until the boundary reaches the ring with lower energy. At this point, the whole ring with low surface energy detaches at once causing a (positive) jump in the normal force. After that, the contact area remains constant and the normal force depends linearly on the indentation depth until this linear dependency reaches the JKR-curve (Fig. 3). Thus, the back movement is very similar to the indentation with the only difference that now the system moves along the JKR curve corresponding to the higher specific surface energy. The hysteresis and the corresponding energy dissipation is solely due to unstable stages (jumps). This mechanism of energy dissipation is very similar to that described by Prandtl [15, 16].

4 Complete Cycle of Attachment and Detachment

The attachment-detachment process becomes especially simple if we assume that the thickness of the rings with different values of specific surface energy are so small that they are not “seen” from the macroscopic point of view. It is easy to give mathematical form to this condition.

In the state (8), the normal force is given by

$$F_2 = 2E^* \int_0^{a_1+h_1} (d_2 - g(\tilde{a})) d\tilde{a} \quad (10)$$

and in the state (9) by

$$F_3 = 2E^* \int_0^{a_1+h_1+h_2} (d_2 - g(\tilde{a})) d\tilde{a}. \quad (11)$$

The jump of the force is estimated as

$$\begin{aligned} F_2 - F_3 &= 2E^* \left[\int_0^{a_2} (d_2 - g(\tilde{a})) d\tilde{a} - \int_0^{a_2+h_2} (d_2 - g(\tilde{a})) d\tilde{a} \right] \\ &= -2E^* \left[\int_{a_2}^{a_2+h_2} (d_2 - g(\tilde{a})) d\tilde{a} \right] \approx 2E^* h_2 \Delta l(a_2) = h_2 \sqrt{8\pi E^* a_2 \gamma_1}. \end{aligned} \quad (12)$$

For a rough estimation let us introduce a “characteristic value” of the specific surface energy, γ (e.g. the average of γ_1 and γ_2), the “characteristic value” of the ring width as h and the “characteristic value” of contact radius, a , e.g. the critical value at the neck formation,

$$a = \left(\frac{9\pi R^2 \gamma}{8E^*} \right)^{1/3}. \quad (13)$$

Then the characteristic value of a jump in the force will be $\Delta F_{\text{Jump}} \approx h \sqrt{8\pi E^* a \gamma}$ and that of the “distance” between the two JKR curves $\Delta F_0 \approx (3/2)\pi R \gamma$, [11]. The condition that the jumps are small compared with ΔF_0 can now be written as

$$\frac{\Delta F_{\text{Jump}}}{\Delta F_0} \approx \frac{h}{R} \sqrt{\frac{32E^* a}{9\pi \gamma}} \ll 1 \quad (14)$$

or

$$h^3 \ll R^2 \frac{\gamma}{E^*}. \quad (15)$$

This form is applicable only for indenters with parabolic shape. Written in the form

$$h \ll a, \quad (16)$$

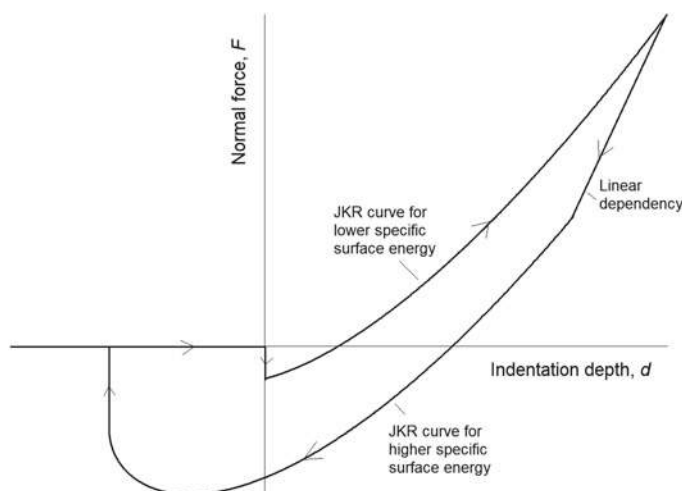


Fig. 4 A complete cycle of the force-indentation dependence for indentation and detachment for the case that the thickness of rings is “microscopic” so that the jumps are not seen on the macroscopic level. In this case, approach occurs along the JKR-curve corresponding to lower specific surface energy and pull-off along the JKR curve for higher surface energy. They are connected by a linear part following the turning point

it can be applied to any shapes. The criterion just means that the characteristic size of heterogeneity should be much smaller than the contact area.

If this condition is fulfilled, we will only see the averaged macroscopic behavior. As was shown in the Sect. 3, the contact configurations and the force-indentation dependencies will follow the JKR solution corresponding to the lower surface energy, γ_1 . On the return, the system follows the JKR curve with higher surface energy, γ_2 . The transition from one curve to the other at the turning point occurs via a linear force-displacement dependency at a constant contact radius (Fig. 4).

Adhesion cycles of the shape qualitatively very similar to that presented in Fig. 4 are often observed in experiment. As an example, in Fig. 5 results are shown, which have been obtained experimentally in the papers [17, 18]. Experiments were carried out by indenting a glass ball against a plane PDMS substrate and subsequent pulling it off. The main features of the behavior are the same as predicted theoretically: Both during loading and during unloading, the system moves along the JKR curves, however, corresponding to different specific surface energies. By turning, a transition from one curve to the other occurs.

Another example of loading-unloading curves showing very clearly the linear transition region after turning from loading to unloading is shown in Fig. 6. The contact area was observed and recorded by a video camera placed beneath the rubber sheet. In the videos (which are not part of this publication), it is clearly seen that after changing the direction of loading the contact area first remained unchanged. During

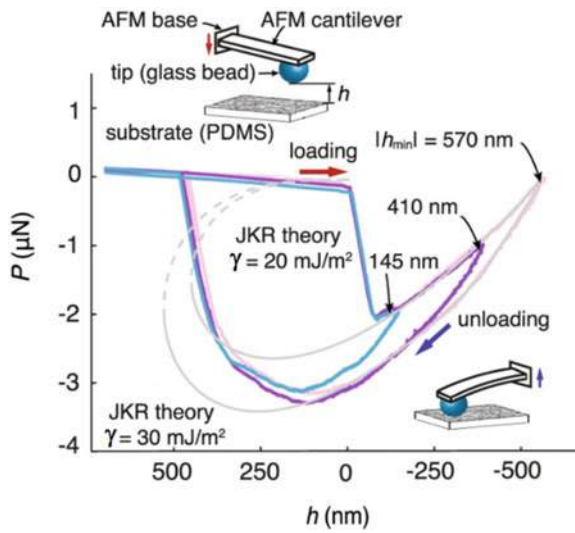


Fig. 5 Experimental loading-unloading curves (adapted from the paper [17]). According to [17], curves were measured using AFM for contacts between a glass sphere and a PDMS substrate. The glass sphere was of diameter $\approx 50\text{ }\mu\text{m}$. The gray dashed curves are the fit of the loading and unloading branches of the measured P - h data to the JKR theory. Comparison with theoretical curves in Fig. 4 shows that the contact behavior in experiment is, at least qualitatively very similar to that predicted theoretically: In the loading phase the system moves along a JKR curve corresponding to a lower specific energy. During unloading the transition from one JKR curve to the other one can be clearly identified

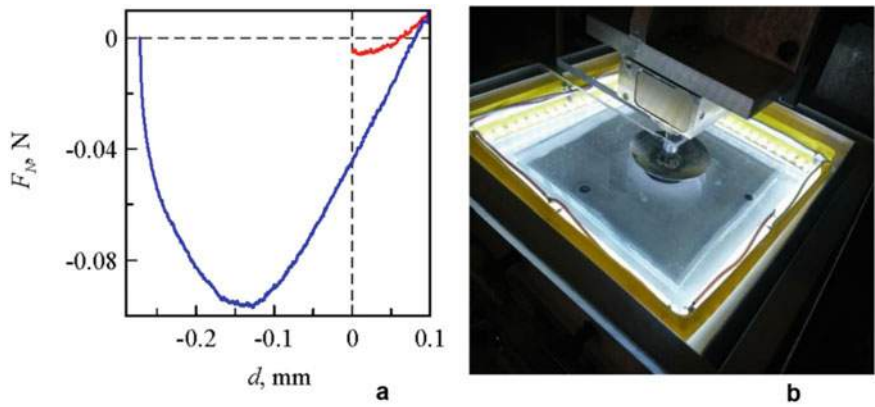


Fig. 6 a Loading-unloading curve for an adhesive contact between a spherical steel indenter (radius of curvature $R = 33\text{ mm}$) and a layer of soft transparent rubber TARNAC CRG N3005 (thickness 5 mm). Part of experimental setup is shown in subplot (b). In the subplot (a), it is clearly seen that after changing the direction of motion, a long linear part of the force-distance dependency is observed. Observation of the contact area via a video camera from beneath the layer shows that during the linear part of this curve the contact area remains constant

this “sticking phase”, the dependency of the normal force on approach follows a linear dependency, which can be easily identified in Fig. 6.

5 Conclusions

We considered a simple adhesive contact with axially symmetric chemical heterogeneity of the interface. In this case, the system follows the JKR curves both during the indentation and detachment phases. However, there exist two different specific surface energies – one for forming the contact (during indentation) and the other one for its destruction (pull off). During the indentation, the system follows the JKR curve corresponding the lower specific surface energy, and during retraction the JKR curve corresponding to the higher value. If the chemical heterogeneity can be considered as “microscopic” (that means that the characteristic wavelength of heterogeneity fulfils the criterion (15) or (16)) this result does not depend on the absolute and relative thicknesses of the regions with different specific surface energies, but depends solely on the values of surface energy itself.

The main conclusions of this paper seem to be very generic. The predicted features are often observed in experimental systems not fulfilling the simple assumptions of the present model. The reason for such generality maybe just that the chemical heterogeneity leads to appearance of a force of friction for the moving contact boundary. From the macroscopic, phenomenological point of view, it is not important what is the physical mechanism leading to microscopic instabilities and thus friction in the boarder line. This can be regular heterogeneity as in the present paper or irregular heterogeneity (which also leads to local instabilities in movement of the contact boundary) or roughness. Macroscopically, the appearance of the force of friction of boundary line is equivalent to existence of two surface energies—for closing and for opening the contact. Thus, the phenomenological appearance may be the same independently of particular mechanism leading to the boundary line friction.

It would be interesting to prove whether this main conclusion will remain valid for non-axially symmetric cases and what are then the governing parameters determining the effective surface energies.

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Theoretical Study of Physico-mechanical Response of Permeable Fluid-Saturated Materials Under Complex Loading Based on the Hybrid Cellular Automaton Method



Andrey V. Dimaki and Evgeny V. Shilko

Abstract We give a brief description of the results obtained by Prof. Sergey G. Psakhie and his colleagues in the field of theoretical studies of mechanical response, including fracture, of permeable fluid-saturated materials. Such materials represent complex systems of interacting solid and liquid phases. Mechanical response of such a medium is determined by processes taking place in each phase as well as their interaction. This raised a need of developing a new theoretical approach of simulation of such media—the method of hybrid cellular automaton that allowed describing stress-strain fields in solid skeleton, transfer of a fluid in crack-pore volume and influence of fluid pressure on the stress state of the solid phase. The new method allowed theoretical estimation of strength of liquid-filled permeable geomaterials under complex loading conditions. Governing parameters controlling strength of samples under uniaxial loading and shear in confined conditions were identified.

Keywords Hybrid cellular automaton · Poroelasticity · Strength · Permeability · Fluid-saturated materials

1 Introduction

Many natural and man-made materials and media, such as permeable rocks (including coal and oil bearing strata) [1–4], bone tissue [5], filtering materials [6], endoprostheses [7] and other, are fluid-saturated porous or cracked porous media. The mechanical response of fluid-saturated permeable materials exhibits some features that differentiate them from solid composite materials in which phases also have different elastic and rheological characteristics. These features are associated with the ability of a liquid/gaseous phase to redistribute in cracks and pores of an enclosing material. As a result of the redistribution, mean stress can either level off in different regions of the material or strongly oscillate due to dilatancy and filling of new discontinuities or due to fluid exchange with the surrounding medium.

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At present, numerous computational methods do exist describing the mechanical behavior of a continuous medium within a certain scale level (finite difference, boundary element, and cellular automata methods [8], etc.). The most popular among them are the finite element method and different variations of the particle method (molecular dynamics, discrete elements, movable cellular automaton (MCA) [9–11] etc.). Distribution of gas and liquid in various porous media is simulated using such modern and intensively developing methods as the lattice Boltzmann method [12] and methods based on the solution of the Navier–Stokes equations on a finite-difference grid. However, the description of a multiscale contrast medium containing interacting solid, liquid, and gas phases within a common approach meets problems connected with description of interrelation between solutions of equations describing the behavior of each phase. Solution of this problem can be found in development of new methods and approaches giving explicit consideration to the multiscale and behavioral peculiarities of studied objects. The most effective among them are methods based on the principles of physical mesomechanics that considers a solid as a multilevel system [13].

Professor Sergey Psakhie founded the method of hybrid cellular automaton (HCA) and was the leader of numerous pioneering works devoted to theoretical study of multiscale and multiphase media using this method. His ideas allowed for solution of many fundamental and practical problems connected with behavior of geological media under complex dynamic loading. Among these media are coal beds filled with gas, porous permeable materials filled with liquid, weakly connected boundaries (shear bands) in geomaterials with gradients of permeability under shear loading conditions.

2 Brief Description of the Hybrid Cellular Automaton Method

One possible theoretical approach to the study of fluid-saturated permeable materials is to use coupled models that account for the following important aspects of such systems behavior: (1) relation of solid skeleton deformation with volume and porosity variation in the pore and crack volume; (2) relation between pore pressure and stress in the enclosing volume of a solid; (3) fluid redistribution in the pore and crack volume. The most famous representatives of this approach are analytical macroscopic models of poroelasticity whose theoretical basis was first discussed by Biot [14, 15]. They were further developed by taking into account a range of scales in real materials, damage accumulation, dilatancy and their influence on the skeleton elastic properties and pore fluid pressure [16–22].

The computational methods based on a discrete representation of the medium are widely used to describe media where fracture on different scales is a factor determining a mechanical response. A well-known representative of this family of models is the discrete element method (DEM) in which the modeled material is

represented as an ensemble of interacting finite-size particles [23–32]. Its major advantages result from the ability of discrete elements to change their surroundings, which is crucial for simulation of complex phenomena such as contact interaction, cracking and fragmentation of solids, flow of granulated media and other.

Various explicit DEMs use different approximations to describe strain distribution within the discrete element volume and the influence of element shape/geometry on its kinematics and interaction with the surroundings [29]. A common approach to description of an element shape is an approximation of an equivalent circular disc or sphere [23, 28, 29]. Further we consider equiaxed or nearly equiaxed elements in approximation of equivalent circular discs and use the term “discrete element” to mention the given simplified representation of an element shape. The given approach has a simple mathematical formulation and apparent advantages in modelling deformation and fracture.

Further development of the DEM formalism expanded its application to a wider range of spatial and structural scales [33–38] and yielded a coupled numerical method for studying permeable media on the meso- and macroscale levels. This coupled numerical method was called the hybrid cellular automaton method. In the HCA method the mechanical response of the enclosing solid is described by the MCA method [39, 40]; fluid filtration and diffusion in cracks and pores of the solids (taken into account implicitly) are described by a finite difference method.

The numerical HCA method is based on separation of the problem into two parts: (1) description of the mechanical behavior of the enclosing solid (skeleton), and (2) description of a fluid transfer in the filtration volume of the solid represented by a system of connected channels, pores, cracks and other discontinuities. Depending on structural features of a considered permeable material and on the simulated scale, the dimensions of discrete elements can be much larger than the linear dimensions of discontinuities in the solid, can be comparable to them or smaller. The influence of “micropores” in the solid skeleton (i.e., pores, channels and other discontinuities whose typical size is smaller than discrete element size) on the mechanical properties and response of a discrete element is accounted for explicitly. Additionally, the MCA method is used to solve the problem of filtration fluid transfer in the network of connected “micropores” of the enclosing solid. Fluid mass transfer between “micropores” and “macropores”, which are considered as regions between spaced and noninteracting discrete elements, is calculated using a finer grid embedded in a laboratory coordinate system (Fig. 1a). The same grid is used to calculate “macropore” volumes (Fig. 1b).

Within the MCA formalism local failure is modelled by changing the state of a pair of interacting elements from linked (or bonded) to unlinked. In simulations described below we apply the modified Drucker-Prager failure criterion taking into account local pore fluid pressure:

$$\sigma_{DP} = 0.5(\lambda + 1)\sigma_{eq} + 1.5(\lambda - 1)(\sigma_{mean} + bP_{pore}) = \sigma_c, \quad (1)$$

where $\lambda = \sigma_c/\sigma_t^i$ is the ratio of compressive σ_c to tensile σ_t strength, σ_{eq} —equivalent stress, b is the dimensionless coefficient determining a contribution of a fluid pore

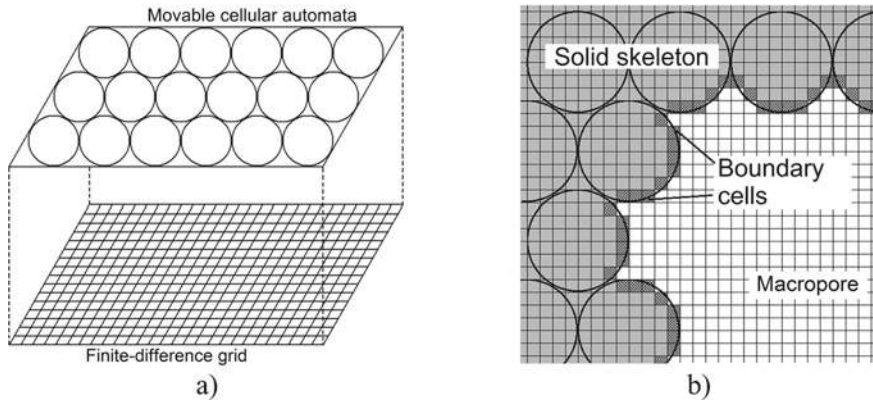


Fig. 1 Layers of discrete elements (movable cellular automata) and finite-difference grid (a); grid cells at a boundary between solid skeleton and macropores (b)

pressure into mean stress σ_{mean} and P_{pore} is the fluid pore pressure. A comprehensive description of details of the numerical implementation of hybrid cellular automaton method is given in [41, 42].

3 Strength of Porous Fluid-Filled Samples Under Uniaxial Loading: A Competition Between Compression and Fluid Filtration

The developed model was applied to study a mechanical response of porous elastic-brittle samples with water-filled filtration volume under uniaxial compression. The samples were fixed between a matrix (at the bottom) and a punch (at the top) that moved downwards and compressed the sample with constant velocity V_y . The compression direction coincided with the vertical sample axis. The problem was solved in a 2D statement in the plane stress approximation. The sample structure was assumed to be homogeneous, without pores and inclusions. The sample height was $H = 0.1$ m and width was $W = 0.05$ m. Numerical experiments were performed using the following parameters of the model material: $K = 37.5$ GPa, $K_s = 107$ GPa, $G = 5.77$ GPa, $\rho = 2000$ kg/m³, $\lambda = 7$, $\sigma_c = 70$ MPa, $K_{fl} = 2.2$ GPa (see details in [41, 42]). Initial value of material porosity was $\varphi_0 = 0.1$.

Two hypotheses of micropore distribution in the solid skeleton were considered with regard to the influence of pore pressure on sample strength.

1. Micropores are distributed homogeneously. Their size is much smaller than the characteristic size of damages and cracks formed during fracture. In this case, we assume that the effect of pore fluid on skeleton strength is governed by the porosity

value and is taken into account by subsequent determination of coefficient b in failure criterion (1): $b = \varphi_0$.

2. Micropores are distributed inhomogeneously, and damages in the material are formed through coalescence of several micropores. In this case, the effect of pore fluid on skeleton strength is directly defined by the pore pressure value: $b = 1$.

The simulation results revealed that pore fluid pressure exerts a strong influence on the mechanical response of brittle porous samples. Other factors that significantly influence the sample strength are the loading rate, characteristic filtration channel diameter (this quantity, along with open porosity, determines material permeability) and geometrical dimensions of the sample.

Fluid pressure in pores of a solid under uniaxial compression is governed by two competing processes: (1) solid skeleton deformation accompanied by pore volume reduction and by pore pressure increase (and, consequently, increasing influence of fluid on the stress state of the skeleton); (2) fluid discharge to the environment through the lateral faces of the sample, due to which pore pressure decreases and the influence of fluid on the stress state of the skeleton also decreases.

A balance of the mentioned two processes is governed by permeability coefficient. At low sample permeability fluid outflow from the sample does not compensate fluid density increase during pore deformation. As a result, fluid pressure in the filtration volume constantly increases in the course of deformation, due to which the effective sample strength decreases. At large permeability, the rate of fluid outflow from the sample is sufficient to reduce pore pressure to zero. The influence of fluid on the stress state of such samples is nearly absent and their strength tends to the strength of a “dry” sample. Between these “limiting” cases the rate of fluid pressure decrease caused by fluid outflow from the sample is comparable to the rate of fluid pressure increase due to solid skeleton compression.

Sample strength is also determined by fluid pressure distribution in the pore volume across the sample cross section. This distribution depends on the ratio of sample width W to its height H . The material volume which is adjacent to the lateral face and through which fluid escapes to the environment decreases with the increasing sample width. Correspondingly, the specific amount and pressure of the fluid retained in pores at the beginning of fracture increases with the growing W/H ratio. This leads to an unexpected conclusion: other factors being equal, the strength of a water-saturated sample of larger width appears to be lower than the strength of a “narrower” sample.

As above, the strength of fluid-saturated permeable samples is governed by the competition of mechanical deformation under applied external load and fluid discharge from the pore space to the environment. To reveal general mechanisms of this competition, we studied the effect of loading rate on uniaxial compressive strength of fluid-saturated samples. Uniaxial compression of fluid-saturated samples with different values of d_{ch} was modeled for different loading rates V_y . The initial pore fluid pressure P_{init} was assumed to be zero (pore volume of samples was completely filled with fluid at atmospheric pressure).

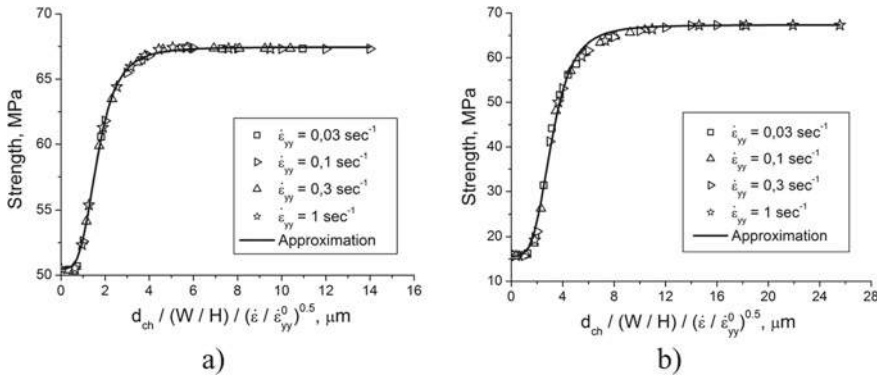


Fig. 2 Generalized dependence of water-saturated sample strength on filtration channel diameter at different loading rates: **a** approximation of homogeneous micropore distribution ($b = \varphi = 0.1$); **b** approximation of relatively large micropores whose evolution leads to macrocrack formation ($b = 1$). Initial fluid pressure in solid skeleton pores is $P_{init} = P_0$. In all calculations the aspect ratio of samples was $W/H = 0.5$

Figure 2 gives the dependences of sample strength on the characteristic filtration channel diameter at different strain rates. The value of parameter b that determines the pore pressure contribution to failure criterion (1) strongly affects the strength of fluid-saturated samples. For example, within the approximation of homogeneous micropore distribution in the solid skeleton ($b = \varphi = 0.1$, Fig. 2a) the pore pressure contribution is rather low (maximum decrease in sample strength does not exceed 25%). Within the second approximation ($b = 1$, Fig. 2b) the strength of water-saturated samples at low permeability values can decrease several-fold. The strength dependence flattens out in this case due to a stronger influence of residual fluid in the pore space on sample strength.

Analysis of the obtained dependences of sample strength on the effective filtration channel diameter d_{ch} at different loading rates and different W/H ratios revealed that they can be reduced to a single dependence of strength on the reduced effective filtration channel diameter:

$$d_{ch} / \left((W/H) \sqrt{\dot{\varepsilon}_{yy} / \dot{\varepsilon}_{yy}^0} \right), \quad (2)$$

where $\dot{\varepsilon}_{yy} = V_y/H$ is the strain rate, and $\dot{\varepsilon}_{yy}^0$ is the scale multiplier that has the dimension of strain rate. The curves shown in Fig. 2 are plotted in these variables ($\dot{\varepsilon}_{yy}^0$ was assumed to be equal to 1 s^{-1}).

It is known that the processes whose occurrence is governed by the competition of several factors or phenomena (e.g., biological population growth, etc.) are often described by a logistic function [43]. Based on the above assumption about the decisive role of the competition between pore pressure increase and fluid outflow from the sample, we used the following logistic function to approximate the dependence of

strength of uniaxially compressed water-saturated samples on the reduced effective filtration channel diameter:

$$\sigma_c(d_{ch}, \dot{\varepsilon}_{yy}) = \sigma_c^{\min} + \frac{\sigma_c^0 - \sigma_c^{\min}}{1 + \left(d_{ch} / \left(d_0 (W/H) \sqrt{\dot{\varepsilon}_{yy} / \dot{\varepsilon}_{yy}^0} \right) \right)^p} \quad (3)$$

where σ_c^0 is the sample strength under uniaxial compression in the absence of fluid in the pore space, σ_c^{\min} is the water-saturated sample strength in the absence of fluid mass transfer, d_0 is the parameter of the approximating function having the dimension of distance, $\dot{\varepsilon}_{yy}$ is the axial strain rate of the sample, and p is the exponent. The parameters of Eq. (3) are defined by the elastic moduli of the solid skeleton and fluid, fluid viscosity, porosity value and so on. As one can see, logistic function (3) allows approximation the numerically calculated data given in Fig. 2 with good accuracy (at $d_0 = 1.62 \mu\text{m}$, $p = 3.7$ for the curve in Fig. 2a and at $d_0 = 3.07 \mu\text{m}$, $p = 4$ for the curve in Fig. 2b).

In order to generalize the results it is useful to take reduced material permeability k as the parameter determining fluid filtration rate:

$$\frac{d_{ch}}{\sqrt{\frac{\dot{\varepsilon}_{yy}}{\dot{\varepsilon}_{yy}^0} \frac{W^2}{H^2}}} \rightarrow \sqrt{\frac{k}{\dot{\varepsilon}_{yy}} \frac{H^2}{W^2} \frac{\dot{\varepsilon}_{yy}^0}{\varphi_0}}. \quad (4)$$

Within this formulation, parameter (2) and approximating function (3) take a more general meaning and can be applied to permeable materials with different structure of filtration volume. Complex relations between the parameters characterizing the mechanical response of the solid skeleton, physical and mechanical properties of fluid and its filtration redistribution dynamics in the system of pores define the nonlinear dependence of sample strength on a combination of these parameters and necessitates the application of numerical methods to study the mechanical response of fluid-saturated porous materials.

4 Influence of Pore Fluid Pressure and Material Dilation on Strength of Shear Bands in Fluid-Saturated Rocks

A range of laboratory and full-scale geological and geophysical research suggests that irreversible deformation in rock samples and rock massifs is strongly localized in shear bands at different scales, the largest of which are tectonic faults [44–46]. These narrow zones not only determine the compliance of rocks in the form of localized relative shear displacements of structural blocks, but control seismic activity of rock massifs. The latter explains the current interest in the mechanical properties of fault zones and the rapid increase in the number of published works in this area.

One of the key mechanical properties is the maximum (or peak) strength of the fault zone under given stress and confinement conditions. Reaching the maximum strength corresponds to a change in the response of the shear band from pervasive strain (strain hardening stage) to strain localization (strain softening stage) [47, 48]. Maximum shear strength estimation of fault zones is both a fundamental and practical problem widely discussed in fault and rock mechanics [49–52].

The conditions for onset of pervasive inelastic strain and subsequent reaching of the maximum strength of shear bands (including faults) are mainly affected by the pore structure and pore fluid pressure. The pore pressure dynamics is controlled by two interrelated processes [53–63]: (1) fluid flow and (2) pore volume change. The pervasive inelastic deformation of rock is often accompanied by its dilatancy [64–66]. The volume of the connected crack-pore space increases during pervasive shear deformation of the shear band, that leads to decrease of local pore pressure. The reduction of fluid pressure reduces the intensity of the relaxation processes associated with the formation of new discontinuities and coalescence of existing ones. This effect is called dilatancy hardening [54]. In turn, fluid inflow is able to compensate for the pore pressure drop and reduce the effect of strain hardening [63, 67]. The ratio of fluid flow rate to strain rate (which governs the dilation rate) determines the specific value of shear strength of shear bands. Note that the influence of the competition between dilatancy and fluid flow on shear strength is strongly pronounced for shear bands surrounded by material blocks with a similar permeability to that of the shear band gouge. This particularly corresponds to healed (consolidated) faults where the difference in the porosity and permeability of the principal slip zone (of width 1–10 cm) and surrounding periphery zone (up to several meters wide) is much less pronounced than in faults with a mature zone of unconsolidated gouge [68].

Conventionally, the effect of pore fluid on the maximum shear strength (hereinafter referred to as strength) of shear bands, including fault zones, has been studied for limiting modes of deformation (very slow and very fast) that correspond to drained and undrained hydrological conditions [69] due to the fact that limiting modes correspond to the long-term creep and short-term dynamic modes of deformation of fault regions. Numerous experimental and theoretical works, starting with the classical work of Brace and Martin [69], show that the strength of permeable rock samples can increase significantly (up to 30–50%) in transition from the drained to undrained condition [67, 70, 71]. This is explained by the limitation of fluid inflow to the increasing pore space of the incipient shear band and corresponding inhibition of pore pressure drop recovery.

There is still no unambiguous understanding of how the strength of a shear band in the depth of constrained permeable rock massif changes in the transition region between the undrained and drained conditions. We studied the nature and functional form of the dependence of the shear band strength on the ratio of shear strain rate to fluid flow rate under constrained conditions corresponding to faults in rock massifs. The study was performed by numerical modelling of the shear deformation of a fluid-saturated permeable shear band using the discrete element method.

Consider a model sample consisting of two blocks separated by an interfacial layer (shear band) in the plane strain approximation (Fig. 3). The blocks mimic regions of

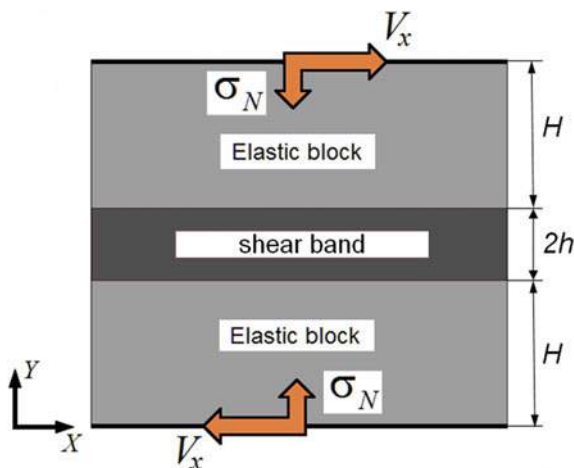


Fig. 3 Sample structure and loading scheme for modelling of constrained shear of a porous fluid-saturated shear band surrounded by porous fluid-saturated blocks

the medium adjacent to the shear band, which are less damaged than the shear band and therefore deform elastically under the considered loading conditions. The shear band of width $2h$ is a layer of an elastic-plastic dilatant material, which simulates the layer of consolidated gouge in principal slip zones of faults [47, 72]. The width of the model blocks was $H = 20h$. We used the following reference values of widths of the shear band and blocks: $2h \approx 1.5$ cm, $H = 15$ cm. The shear band and blocks were assumed to be permeable and fluid saturated.

The model shear band with surrounding fragments of blocks was numerically simulated by the discrete element method using a fully coupled macroscopic model of fluid-saturated porous brittle materials [33–41]. Within the model discrete elements simulating parts of the shear band and surrounding blocks are treated as porous and permeable. The effect of the fluid contained in the crack-pore volume of a discrete element on its stress state is described based on Biot linear model of poroelasticity [14, 15]. The inelastic behavior of the permeable brittle material of the discrete element is described using a plastic flow model of rocks with a non-associated flow law and the Mises–Schleicher yield criterion (Nikolaevsky’s model) [16]. The elastic characteristics of the shear band and blocks were assumed to be similar and corresponded to typical values for sandstones with a porosity of 10–15% (Young’s modulus $E = 15$ GPa, Poisson’s ratio $\nu = 0.3$). The material of discrete elements modelling the blocks was treated as elastic-brittle and high-strength. The material of discrete elements modelling the shear band was a model elastic-plastic material with linear hardening with the following plasticity and strength parameters: $\beta = 0.57$, $\Lambda = 0.36$, $Y = 10.84$ MPa (this corresponds to a yield stress of 28 MPa under uniaxial compression), strain hardening modulus $\Pi = 515$ MPa, uniaxial compressive strength $\sigma_c = 40$ MPa, uniaxial tensile strength $\sigma_t = 13.33$ MPa ($\lambda = 3$). The

calculations were carried out at an initial mean stresses σ_{mean}^0 below the brittle-ductile transition threshold (in this case, at $\sigma_{mean}^0 < 40$ MPa).

The initial values of porosity ($\phi_0 = 0.1$) and permeability k_0 of the shear band and the blocks were assumed to be equal. This approximation is consistent with the low gradients of porosity and permeability in central zones of healed (consolidated) faults. Initially, all interacting elements were linked that simulates a consolidated shear band.

We modelled constrained shear of the sample in the horizontal plane along the X axis (Fig. 3). Periodic boundary conditions were specified on the lateral faces in the horizontal direction to simulate an infinitely long shear band. The sample was loaded in two stages. At the first stage, a normal load σ_N was applied to the upper and lower sample faces. The initial fluid concentration in the pore space of the sample was chosen so as to create the specified pore pressure P_{pore}^0 in the sample. There was no plastic deformation in the sample by the end of the first loading stage. The stress and pore pressure distributions were homogeneous. At the second stage, the sample was subject to simple shear by applying constant tangential velocity V_x and zero normal velocity (along the Y axis) to the upper and lower faces to fulfil the constrained shear condition. The sample deformation proceeded until crack initiation in the shear band.

The described 2D system models a horizontal cross section of a healed fault between structural blocks of a rock massif at a certain depth. Note that the initial “horizontal” (in the XY plane) stresses in the given formulation of the problem exceed the “vertical” ones. This is consistent with experimental data indicating that horizontal stresses are considerably higher than vertical ones in regions with high deformation activity [73]. We used isolated conditions on the external surfaces of the sample (hydraulically isolated sample) that correspond to the hydrological conditions in the central regions of fault zones in the bulk of low permeable host rocks.

The simulation results showed that at high strain rates the magnitude of shear strength tends toward the upper limit, and at low strain rates—toward the minimal value. Such regularity was first observed by Brace and Martin [69], and then reported in numerous experimental and theoretical studies of confined compression of rock samples and shear loading of model fault zones (see e.g. [67]).

The reduction of shear strength from the upper to the lower limit with reduction of strain rate is not monotonous. At a certain intermediate strain rate the shear strength reaches a local minimum. Further reduction of strain rate leads to an *increase* of shear strength up to a local maximum. At even smaller strain rates the shear strength again *decreases* down to the lower limit. This result quantitatively agrees with recent experimental studies [74–76].

We varied the shear strain rate $\dot{\epsilon}_{xy} = V_x/(H + h)$, the initial permeability of the blocks and the shear band k_0 (at fixed $\phi_0 = 0.1$), the dynamic fluid viscosity η , and system size ($2H + 2h$) within several orders of magnitude: $\dot{\epsilon}_{xy}$ from $5 \times 10^{-4} \text{ s}^{-1}$ to 1 s^{-1} , k_0 from 10^{-18} m^2 to 10^{-13} m^2 , η from $2 \times 10^{-4} \text{ Pa s}$ to $2 \times 10^{-2} \text{ Pa s}$ (dynamic viscosity of water at room temperature is about 10^{-3} Pa s), ($2H + 2h$) from 15 to 150 cm. We found that the parameter combination

$$A_{xy} = \frac{\dot{\varepsilon}_{xy} \eta (H + h)^2}{k_0} \quad (5)$$

unambiguously determines the value of shear strength of the shear band for a given initial mean stress σ_{mean}^0 , pore pressure P_{pore}^0 , and ratio h/H . In other words, shear band zones have the same shear strength if they are characterized by the same value of A_{xy} , (even if the specific values of the parameters k_0 , η , $\dot{\varepsilon}_{xy}$, and h differ by orders of magnitude). The parameter A_{xy} means the relation of strain rate to fluid flow rate.

Figure 4 shows a typical dependence of the shear strength τ_c of the modelled shear band on the parameter A_{xy} for a hydraulically isolated system. Each point of the curve corresponds to a separate calculation at given values of k_0 , η , $\dot{\varepsilon}_{xy}$ and h (at $h/H = const$, $\sigma_N = const$, $P_{pore}^0 = const$). The region $A_{xy} \rightarrow \infty$ (region I in Fig. 4) corresponds to combinations of k , η , $\dot{\varepsilon}_{xy}$ and h where the fluid flow rate is extremely low compared to the rate of pore pressure change caused by pore volume variation. This corresponds to the hydrological conditions close to the undrained condition of the shear band. The region of low A_{xy} values (region III in Fig. 4) corresponds to low shear rate, low fluid viscosity or high permeability of the blocks. In this region the fluid flow rate is relatively high, and the hydrological conditions for the shear band approach a fully drained (the pore pressure distribution in the sample is close to homogeneous during the entire course of deformation).

The curve shown in Fig. 4 has three characteristic regions where the change of strength is monotonous, that implies a presence of a dominant mechanism determining the direction of the change.

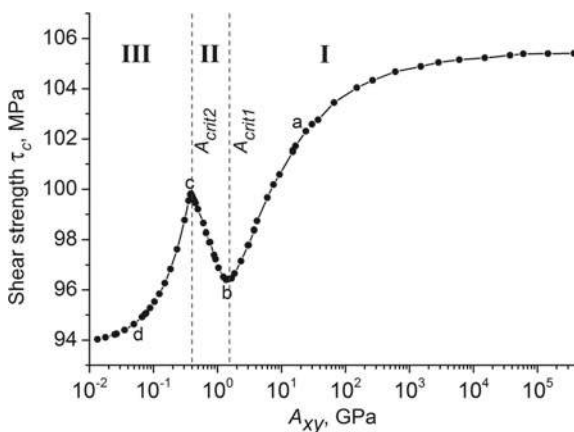


Fig. 4 A typical dependence of the shear strength τ_c of the shear band on the parameter A_{xy} for a hydraulically isolated sample. Roman numerals I–III mark the curve regions corresponding to different behaviour modes of the fluid-saturated sample under shear loading. The values A_{crit1} and A_{crit2} correspond to the local minimum and maximum shear strength. The top and bottom faces of the sample are fixed in vertical direction. Lower A_{xy} imply faster fluid flow or lower strain rate, higher A_{xy} imply slower fluid flow or higher strain rate

In region I the dominating mechanism lies in the decrease of the linear dimensions of the blocks due to fluid outflow to the shear band (poroelastic contraction). Decrease in the value of A_{xy} is accompanied by the inflow of a large amount of fluid into the shear band and hence by reduction of the constraint imposed on the shear band by the compressed blocks (effective normal stiffness of the blocks decreases). This mechanism determines the decrease of the shear strength in region I as A_{xy} decreases.

In region II the trend-determining mechanism is tied to the increase of the dilation rate of the shear band with decreasing value of A_{xy} due to slowing of pore pressure reduction and maintaining nonzero pore pressure during most of the shear process. This mechanism provides an increase in the absolute value of effective mean stress in the sample and hence increase in the strength of the shear band in region II as A_{xy} decreases.

In region III the trend-determining mechanism is linked to the fact that pore pressure in the shear band remains non-zero during the entire shear process. In this region pore pressure in the shear band is higher in the samples characterized by lower values of A_{xy} . Because of this fact an absolute value of effective mean stress in a shear band is also lower in the samples with lower A_{xy} . Decrease of an absolute value of effective mean stress leads to gradual decrease in the shear strength down to the absolute minimum at $A_{xy} \rightarrow 0$.

The described three parts of the curve $\tau_c(A_{xy})$ have sigmoid profiles that is the result of the competition between shear band dilatancy and poroelastic contraction of the blocks due to fluid outflow. Analysis of the obtained result allowed formulating the following general dependence of the shear strength of constrained shear band zones on the parameter A_{xy} . The dependence is expressed as a sum of a constant and three sigmoid contributions:

$$\tau_c = \tau_0 + \frac{\tau_1}{1 + (c_1 A_{xy})^{-p_1}} + \frac{\tau_2}{1 + (c_2 A_{xy})^{p_2}} - \frac{\tau_3}{1 + (c_3 A_{xy})^{p_3}}, \quad (6)$$

where τ_1, τ_2 and τ_3 are the amplitudes of contributions of the three above-mentioned mechanisms to the shear strength, c_1, c_2 and c_3 are the inverse positions of the sigmoid midpoints, the exponents p_1, p_2 and p_3 determine the steepness of the sigmoid functions and τ_0 is a constant contribution independent of the fluid flow dynamics.

The amplitudes of the contributions have the following physical meanings. The constant contribution τ_0 is the strength of the shear band in the absence of fluid flow (or under high strain rate) at a constant value of normal force σ_N applied to the upper and lower sample faces during the entire course of shear. Another three contributions are concerned with squeezing of the blocks due to shear band dilation in a mechanically constrained sample.

The specific parameter values of Eq. (6) depend on mechanical characteristics of the skeleton of shear band and blocks, bulk modulus of fluid, the h/H ratio, initial mean stress σ_{mean}^0 , fluid content in the blocks and hydrological conditions.

5 Conclusion

We have presented the results of application of the hybrid numerical technique for theoretical study of deformation and fracture of fluid-saturated permeable materials and media. In the framework of this technique the simulated medium is considered as a superposition of two interdependent layers. One layer is represented by an ensemble of particles (simply deformable discrete elements), and the other—by a finite-difference grid. This approach combined with Biot's model of poroelasticity is suitable for studying complex and interrelated processes of solid skeleton deformation and fracture and fluid redistribution (mass transfer) in the pore volume.

Strength of water-saturated elastic-brittle samples under uniaxial compression significantly depends on fluid pressure in the pore space as well as on solid permeability, physical and mechanical properties of fluid, strain rate and sample dimensions. The numerically simulated dependences of strength of water-saturated samples are well described by a logistic function. This bears witness to the decisive role of the competition of two processes, such as pore pressure increase and fluid outflow to the environment, in permeable brittle materials in the course of mechanical loading.

It was shown that strength of a saturated shear band is directly connected with shear rate, fluid viscosity and permeability of the shear band zone and surrounding massif. The governing combination of these parameters A_{xy} , together with the obtained empirical dependence $\tau_c(A_{xy})$ allows prediction of shear band strength under given loading conditions. The latter may be especially important for estimation of transition point of the shear mode of consolidated fault segments from stick to dynamic slip.

The reported results demonstrate broad potentials of the developed DEM-based coupled model of a poroelastic medium and show the importance of numerical modeling application to the study of the mechanical properties (including strength) of dynamically loaded fluid-saturated materials.

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Transfer of a Biological Fluid Through a Porous Wall of a Capillary



Nelli N. Nazarenko and Anna G. Knyazeva

Abstract The treatise proposes a model of biological fluid transfer in a dedicated macropore with microporous walls. The distribution of concentrations and velocity studies in the capillary wall for two flow regimes—convective and diffusive. The largest impact on the redistribution of concentration between the capillary volume and its porous wall is made by Darcy number and correlation of diffusion coefficients and concentration expansion. The velocity in the interface vicinity increases with rising pressure in the capillary volume or under decreasing porosity or without consideration of the concentration expansion.

Keywords Capillary · Diffusion · Peclet number · Convective and diffusive flow regimes

1 Introduction

Contemporary medicine widely implements agents for tissue culture, delivery systems for pharmaceuticals, implants, bandages, arterial conduits, etc. The efficacy of all synthesized materials depends on their structure, including the structure of the pore space, which largely controls the kinetics of biochemical processes. For example, an implant should possess a strictly determined pore size promoting the formation of blood vessels during tissue growth. The structure of biological porous media is multiscale. Along with macroscopic pores, there are a lot of capillaries. Pore walls, in turn, consist of several layers, each of which has its own structure. The properties of the surfaces of pores and capillaries also affect the flow of biological fluids.

The main biological fluids of a person include blood, tissue fluid and lymph. The first performs mainly a transport function. Essentials for life enter the cells through the tissue fluid from the blood into the cells. The main function of the third fluid

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is protective. Lymph destroys pathogens and ensures the return of tissue fluid into the bloodstream. The blood vessels through which blood flows from the heart form the arterial system, and the vessels that collect blood and carry it to the heart form the venous system. The metabolism between the blood and body tissues is carried out using capillaries that penetrate the organs and most tissues. The main functions of the blood and circulatory system are to connect organs and cells to ensure their vital functions—in the delivery of oxygen, nutrients, hormones, excretion of decomposition products, maintaining a constant body temperature, and protection from harmful microbes [1, 2]. All this suggests the need to study the flow of biological fluids in the system of vessels and capillaries, taking into account the features of the structure and rheological properties and the development of appropriate models [3]. The rheological properties of blood are mainly due to the processes of hydrodynamic interaction of erythrocytes with plasma, which contribute to the formation and decay of aggregates, rotation and deformation of red blood cells, their redistribution, and the corresponding orientation in the flow [4]. Blood is a heterogeneous and multiphase physical and chemical system. It can be represented as a suspension and non-Newtonian fluid with complex rheological properties. In addition to modeling blood flow in large blood vessels [2, 4–8] there are a number of works in the literature in which the flow in capillaries is simulated.

For example, work [9] has analyzed three variants of mathematical models describing the flow of a viscous incompressible fluid in a long cylindrical capillary with its internal surface covered by a permeable porous layer. The authors have shown that for thin weakly permeable porous layers on the capillary walls, the Brinkman model is not applicable; one just can use the Navier slip condition. If the porous layer is thick and/or is weakly permeable, it is not allowed to neglect the effect of the flow in it on the total fluid flow rate through the capillary, and an adequate description of the filtration process should be made using the Brinkman model.

In [10, 11], the authors have studied the blood flow through porous blood vessels taking into account an electromagnetic field. They have suggested a blood flow model in an artery with porous walls within the model of a non-Newtonian fluid in the presence of electromagnetic field. In these works, the viscosity of the non-Newtonian fluid depended on the temperature and magnetic field and was calculated by the models of Reynolds and Vogel or was assumed to be constant.

There are a lot of works devoted to modeling of blood flow in capillaries [12–14]. The authors investigate the influence of diverse parameters on the fluid motion in capillaries: vessel curvature [15], capillary radius and shape [16–18], dynamics of oxygen transportation [19], hemodynamics of vascular prostheses and implants [20, 21].

Important applications of biomedical systems, such as biological tissues, require taking into account the flow, heat and mass exchange through a porous medium [22]. The theory of transfer in a porous medium on the basis of various models, such as Darcy-Brinkman model of momentum transfer and local thermal equilibrium for energy transfer, were analyzed by the authors and can be particularly useful in describing different biological applications.

In general case, biological fluids possess specific rheological properties. As a rule, biological fluids are non-Newtonian fluids that are described by various rheological models. In the literature, models of viscoelastic, viscoplastic, pseudo-plastic and dilatant fluids are widely used. All the models, with due consideration of complex rheology, are non-linear. The non-linear effects also manifest when accounting the dependence of properties (for instance, viscosity) on concentration.

Current work suggests a model of biological fluid transfer in a selected macropore with microporous walls. Unlike [10, 11], we assume isothermal conditions; however, we assume some the state equations for pressure in the fluid to be differential [23, 24], which yields a non-linear coupled model.

2 General Equations

Let us formulate the problem on the transfer of a biological fluid (or a pharmaceutical) in a selected cylindrical macropore with radius R_1 having microporous walls (Fig. 1). Area 1 is the macropore, area 2 is the porous layer with thickness $\delta = R_2 - R_1$.

To construct the model we use the continuity equation:

$$\frac{\partial \rho}{\partial t} + \nabla \cdot (\rho \mathbf{v}) = 0, \quad (1)$$

balance equation for species

$$\rho \left(\frac{\partial C_k}{\partial t} + \mathbf{v} \nabla C_k \right) = -\nabla \cdot \mathbf{J}_k, \quad (2)$$

and motion equation

$$\rho \frac{\partial \mathbf{v}}{\partial t} + \rho \mathbf{v} \nabla (\mathbf{v}) = -\nabla \cdot \boldsymbol{\sigma} + \rho \mathbf{F}, \quad (3)$$

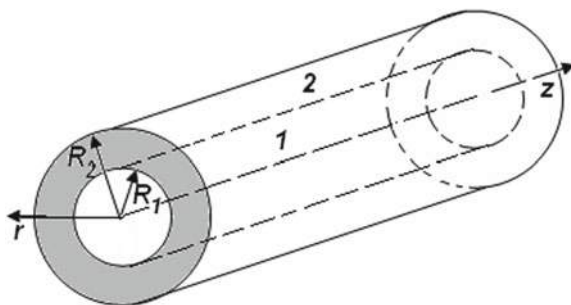


Fig. 1 Cylindrical pore with radius R_1 having porous walls

where ρ is the density, \mathbf{v} is the velocity of centre of mass; C_k —species (component) concentrations; \mathbf{J}_k is diffusion flux of this component; $\boldsymbol{\sigma}$ is stress tensor; \mathbf{F} is the mass force vector; $\nabla \dots \equiv \text{grad} \dots$; $\nabla \dots \equiv \text{div} \dots$.

We will describe the flow in the macropore (area 1) using Navier-Stokes equations. The microporous medium (area 2) will be modeled as Brinkman medium. In a first approximation, the biological fluid is assumed incompressible. Navier-Stokes equations follows from (3) when

$$\sigma_{ij} = -P\delta_{ij} + 2\mu e_{ij}, \quad (4)$$

and $\mathbf{F} = \mathbf{F}_1 = -\frac{1}{\rho}\nabla(gz)$. Here p is hydrodynamic pressure and e_{ij} is the tensor of strain rates,

$$e_{ij} = \frac{1}{2} \left(\frac{\partial V_i}{\partial x_j} + \frac{\partial V_j}{\partial x_i} \right),$$

V_i are components of the velocity vector.

Brinkman medium appears when we assume

$$\mathbf{F} = \mathbf{F}_1 + \mathbf{F}_2, \quad (5)$$

where \mathbf{F}_2 is the force of internal friction depending on filtration velocity, \mathbf{w} . Then $\mathbf{v} = \mathbf{w}/a$, $a = S_p/S$, and S_p is the area occupied by pores in the section S .

If the fluid is incompressible (which is usually accepted for slow flows), instead continuity equations (1) will remain:

$$\nabla \cdot \mathbf{v}_i = 0; \quad i = 1, 2. \quad (6)$$

As a result we obtain for area inside the capillary:

$$0 < r < R_1: \quad \rho_1 \left(\frac{\partial C_k}{\partial t} + \mathbf{v}_1 \nabla C_k \right) = -\nabla \cdot \mathbf{J}_k, \quad (7)$$

$$\rho_1 \left(\frac{\partial \mathbf{v}_1}{\partial t} + \mathbf{v}_1 \nabla \mathbf{v}_1 \right) = -\nabla p_1 + \rho_1 gz + \nabla \cdot (\mu_1 \nabla \mathbf{v}_1), \quad (8)$$

and for the porous walls:

$$R_1 < r < R_2: \quad \rho_2 \left(\frac{\partial C_k}{\partial t} + \mathbf{v}_2 \nabla C_k \right) = -\nabla \cdot \mathbf{J}_k, \quad (9)$$

$$\rho_2 \left(\frac{\partial \mathbf{v}_2}{\partial t} + \mathbf{v}_2 \nabla \mathbf{v}_2 \right) = -\nabla p_2 + \rho_2 gz + \left(\nabla \cdot (\mu_2 \nabla \mathbf{v}_2) - \mu_2' m \frac{\mathbf{v}_2}{k_f} \right). \quad (10)$$

Here $\mathbf{v}_1, \mathbf{v}_2, \rho_1, \rho_2$ are the vectors of velocities and densities of the liquid in areas 1 and 2, C_k is the concentration of the k -th component, $\mathbf{J}_k = -D_k \rho_i \nabla C_k$ is the diffusion flux of the k -th component; P_1, P_2, μ_1, μ_2 are the pressure and viscosity of the fluid in areas 1 and 2, μ_2' is the viscosity in the Darcy's law, in general case it differs from μ_2 ; g is the force of gravity; k_f is the permeability of the porous medium; m is the porosity of pore walls.

In the case of slow (crawl) flow, the second summands in the left brackets of the motion equations for porous walls can be neglected.

We should add the state equation connecting the pressure with temperature and fluid composition. For constant temperature, we can write [23, 25]

$$dp = -\rho \beta_T^{-1} d\gamma + \sum_{k=1}^n p_k dC_k \quad (11)$$

where $p_k = \alpha_k \beta_T^{-1}$, α_k is concentration expansion coefficients, β_T is isothermal compressibility coefficient, $\beta_T^{-1} = K$; K is bulk module for fluid. Then for incompressible fluid with constant properties for each area we have

$$p_2 - p_{20} = 3K\alpha(C_2 - C_{20}) \quad \text{and} \quad p_1 - p_{10} = 3K\alpha(C_1 - C_{10}), \quad (12)$$

where C_{10} and C_{20} is preset zero approximation, p_{10} and p_{20} —is initial pressures values in areas.

3 Stationary Model

From (7)–(10) we obtain stationary model for individual pore. The hydrodynamic part of the problem will include equations:

$$0 < r < R_1 : \quad \rho_1 V_1 \frac{dV_1}{dr} = -\frac{dp_1}{dr} + \frac{1}{r} \frac{d}{dr} \left(r \mu_1(C_1) \frac{dV_1}{dr} \right), \quad (13)$$

$$R_1 < r < R_2 : \quad \frac{dp_2}{dr} - \left(\frac{1}{r} \frac{d}{dr} \left(r \mu_2(C_2) \frac{dV_2}{dr} \right) - \mu_2'(C_2) m \frac{V_2}{k_f} \right) = 0, \quad (14)$$

where $V_k, k = 1, 2$ are radial components of velocity for areas.

The boundary conditions for the hydrodynamic part of the problem will be as follow. In the point $r = 0$ we have the symmetry condition

$$V_1 = 0. \quad (15)$$

In the interface between two areas, mass velocities and stress tensor components are equal

$$r = R_1 : \quad \rho_1 V_1 = m \rho_2 V_2, \quad -p_1 + \mu_1(C_1) \frac{dV_1}{dr} = m \left(-p_2 + \mu_2(C_2) \frac{dV_2}{dr} \right). \quad (16)$$

We can assume that the outer wall of the capillary ($r = R_2$) is free of load, on

$$\sigma_{rr} = -p_2 + \mu_2 \frac{dV_2}{dr} = 0 \quad (17)$$

or (other case)

$$V_2 = 0. \quad (18)$$

The viscosities linearly depend on concentration:

$$\mu_1(C_1) = \mu_{10} + \mu_{11}C_1, \quad \mu_2(C_2) = \mu_{20} + \mu_{21}C_2. \quad (19)$$

For the diffusion part of the problem, we have:

$$0 < r < R_1 : \quad \rho_1 V_1 \frac{dC_1}{dr} = \frac{1}{r} \frac{d}{dr} \left(D_1 \rho_1 r \frac{dC_1}{dr} \right), \quad (20)$$

$$R_1 < r < R_2 : \quad \rho_2 V_2 \frac{dC_2}{dr} = \frac{1}{r} \frac{d}{dr} \left(D_2 \rho_2 r \frac{dC_2}{dr} \right), \quad (21)$$

$$r = 0 : \quad \frac{dC_1}{dr} = 0, \quad (22)$$

$$r = R_1 : \quad C_1 = m C_2, \quad \rho_1 \left[D_1 \frac{dC_1}{dr} - V_1 C_1 \right] = \rho_2 m \left[D_2 \frac{dC_2}{dr} - V_2 C_2 \right], \quad (23)$$

$$r = R_2 : \quad \rho_2 \left(D_2 \frac{dC_2}{dr} - V_2 C_2 \right) = \Omega, \quad (24)$$

where D_1, D_2 are the diffusion coefficients in areas 1 and 2.

The condition (22) is symmetry condition; first of (23) follows from chemical potential continuity, second of (23) is the equality of the total mass flows; condition (24) contains the mass sink on the outer wall of the capillary Ω .

Taking into account the connection between pressure and concentration (12), from (13), (14), (16), (17) we obtain

$$\begin{aligned} \rho_1 V_1 \frac{dV_1}{dr} &= -3K\alpha \frac{dC_1}{dr} + \frac{1}{r} \frac{d}{dr} \left(r \mu_1(C_1) \frac{dV_1}{dr} \right); \\ 3K\alpha \frac{dC_2}{dr} - \left(\frac{1}{r} \frac{d}{dr} \left(r \mu_2(C_2) \frac{dV_2}{dr} \right) - \mu_2'(C_2) m \frac{V_2}{k_f} \right) &= 0; \end{aligned}$$

$$\begin{aligned}
& - (3K\alpha(C_1 - C_{10}) + p_{10}) + \mu_1(C_1) \frac{dV_1}{dr} \\
& = m \left(- (3K\alpha(C_2 - C_{20}) + p_{20}) + \mu_2(C_2) \frac{dV_2}{dr} \right); \\
& \sigma_{rr} = - (3K\alpha(C_2 - C_{20}) + p_{20}) + \mu \frac{dV_2}{dr} = 0.
\end{aligned}$$

In this model we assumed that viscosity of fluid and diffusion coefficients in pore and in porous wall are different, that connect with special structure of porous space affecting the fluid mobility. These problem is coupling in general case.

4 Special Case

The simplest stationary diffusion model for individual pore can be analyzed for the case when the pressure gradient along the macro pores is given, and the fluid composition in pore is fixed (we neglect the gravitational force):

$$\nabla p_1 = \omega = \text{const}, \quad C_1 = C_{10}. \quad (25)$$

Because in interface $\rho_1 V_1 = m \rho_2 V_2$ and the pressure is proportional to concentration, then we do not mistake if assume:

$$\nabla p_1 = \text{const} = \beta \omega \sim \nabla p_2, \quad (26)$$

where ω , β is some constants. In this case the hydrodynamical part of the problem turns to

$$\omega - \frac{1}{r} \frac{d}{dr} \left(r \mu_1(C_{10}) \frac{dV_1}{dr} \right) = 0; \quad (27)$$

$$\omega \beta - \left(\frac{1}{r} \frac{d}{dr} \left(r \mu_2(C_2) \frac{dV_2}{dr} \right) - \frac{V_2}{k_f} \right) = 0; \quad (28)$$

$$r = 0 : \quad V_1 = 0; \quad r = R_2; \quad V_2 = 0; \quad (29)$$

$$r = R_1 : \quad \rho_1 V_1 = m \rho_2 V_2; \quad \mu_1(C_{i0}) \frac{dV_1}{dr} = m \mu_2(C_2) \frac{dV_2}{dr}. \quad (30)$$

For the case of constant viscosity, the exact analytical solution of this problem is presented in [26].

Diffusion part of the problem takes a place only for the area $R_1 \leq r \leq R_2$:

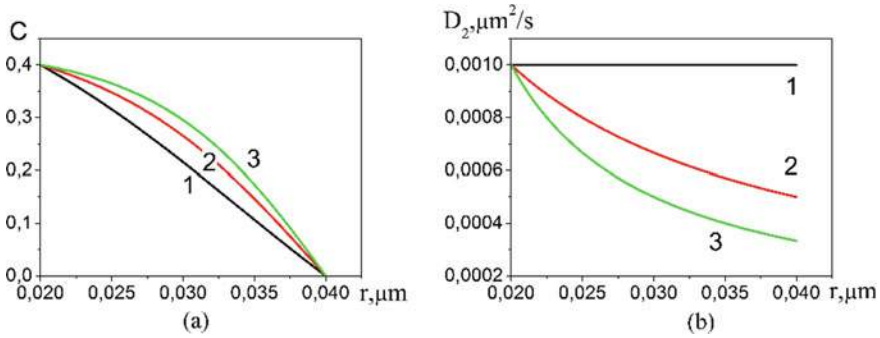


Fig. 2 Concentration distribution in the wall of pore (a) for different diffusion coefficients (b). $D_{20} = 10^{-3} \mu\text{m}^2/\text{s}$, 1— $\alpha_{p2} = 0$; 2— $\alpha_{p2} = 6$; 3— $\alpha_{p2} = 10$

$$\rho_2 V_{r2}(r; C_{10}, C_2, m, \omega) \frac{dC_2}{dr} = \frac{1}{r} \frac{d}{dr} \left(D_2 \rho_2 r \frac{dC_2}{dr} \right);$$

$$r = R_1 : C_{10} = m C_2;$$

$$r = R_2 : C_2 = 0.$$

We assume that velocity distribution in the walls is given and does not depend on concentration. It is obviously, when the velocity is equals to zero, and diffusion coefficient is constant value, we come to concentration distribution coinciding with the exact analytical solution (Fig. 2a, b—lines 1):

$$C_2 = \frac{C_{10}}{m} \frac{\ln(r/R_2)}{\ln(R_1/R_2)}.$$

This solution does not contain density and diffusion coefficient.

If diffusion coefficient depends on space coordinate (that could be connected with the change of pore structure, for example, using the equation $D_2 = D_{20}/(1 + \alpha_{p2}(r - R_1))$), then concentration distribution changes (2 and 3 curves correspondingly) in this figure.

The positive value of given filtration velocity effects on concentration distribution similarly (Fig. 3). However, the type of the velocity distribution is not essential concentration (Fig. 4).

The concentration distribution in Fig. 4b is given for the velocity functions $V_2(r)$, presented in the Fig. 4a. The change of the velocity with the coordinate could be connects with the complex structure of porous space, with structure of pore surface, with their specific tortuosity, the close pores availability leading to inhibition of concentration distribution along pore walls.

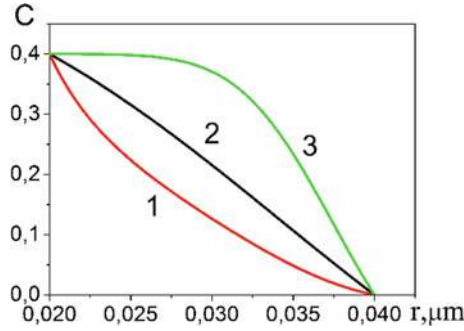


Fig. 3 Concentration distribution for given filtration rate. 1— $V_2 = 0.1$; 2— $V_2 = 0$; 3— $V_2 = 0.2$ $\mu\text{m/s}$, $D_{20} = 10^{-3} \mu\text{m/s}$

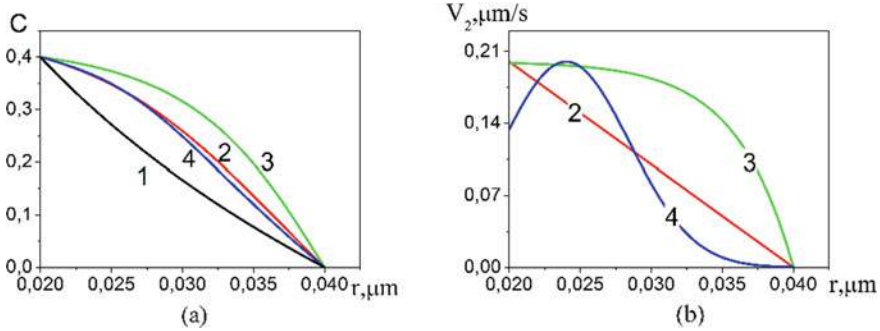


Fig. 4 Concentration distribution (a) and liquid velocity in pore wall (b). Black line—is exact analytical solution for $V_2 = 0$; the colors of the lines to the left correspond to colors of the lines to the right. $D_{20} = 10^{-3} \mu\text{m/s}$

5 Dimensionless Variables and Parameters in Total Stationary Model

Let us introduce the following dimensionless variables:

$$\xi = \frac{r}{R_2}, \quad \bar{V}_i = \frac{V_i}{\mu_{10}/\rho_1 R_2}, \quad \bar{p}_i = \frac{p_i}{\mu_{10}^2/\rho_1 R_2^2}.$$

Then the equations and boundary conditions in dimensionless variables will be rewritten as

$$0 < \xi < 1 - \Delta: \quad \bar{V}_{r1} \frac{d\bar{V}_1}{d\xi} = -\frac{d\bar{p}_1}{d\xi} + \frac{1}{\xi} \frac{d}{d\xi} \left(\xi \bar{\mu}_1(C_1) \frac{d\bar{V}_1}{d\xi} \right), \quad (31)$$

$$Pe_D \bar{V}_1 \frac{dC_1}{d\xi} = \frac{1}{\xi} \frac{d}{d\xi} \left(\xi \frac{dC_1}{d\xi} \right), \quad (32)$$

$$1 - \Delta < \xi < 1: \quad Da \frac{d\bar{p}_2}{d\xi} - \left(\frac{Da}{\xi} \frac{d}{d\xi} \left(\xi \bar{\mu}_2(C_2) \frac{d\bar{V}_2}{d\xi} \right) - \bar{\mu}'_2 m(C_2) \bar{V}_2 \right) = 0, \quad (33)$$

$$Pe_D \bar{V}_2 \frac{dC_2}{d\xi} = \bar{D} \frac{1}{\xi} \frac{d}{d\xi} \left(\xi \frac{dC_2}{d\xi} \right), \quad (34)$$

$$\xi = 0: \quad \bar{V}_1 = 0, \quad \frac{dC_1}{d\xi} = 0, \quad (35)$$

$$\xi = 1 - \Delta: \quad \bar{V}_{r1} = m\bar{\rho}\bar{V}_{r2}, \quad -\bar{p}_1 + \bar{\mu}_1(C_1) \frac{d\bar{V}_{r1}}{d\xi} = m \left(-\bar{p}_2 + \bar{\mu}_2(C_2) \frac{d\bar{V}_{r2}}{d\xi} \right), \quad (36)$$

$$C_1 = mC_2, \quad \frac{dC_1}{d\xi} - Pe_D \bar{V}_1 C_1 = m\bar{\rho} \left(\bar{D} \frac{dC_2}{d\xi} - Pe_D \bar{V}_2 C_2 \right), \quad (37)$$

$$\xi = 1: \quad \bar{\mu}_2 \frac{d\bar{V}_{r2}}{d\xi} = \bar{p}_{20} + K_{bez}(C_2 - C_{20}), \quad (38)$$

$$\bar{D} \frac{dC_2}{d\xi} - Pe_D \bar{V}_2 C_2 = \bar{\Omega}, \quad (39)$$

where $\bar{\mu}_1(C_1) = 1 + \alpha_1 C_1$, $\bar{\mu}_2(C_2) = \beta + \alpha_2 C_2$, $\bar{\mu}'_2 = \bar{\mu}_2$, $\bar{p}_2 - \bar{p}_{20} = K_{bez}(C_2 - C_{20})$ and $\bar{p}_1 - \bar{p}_{10} = K_{bez}(C_1 - C_{10})$.

Stationary model contains following dimensionless parameters:

$$Pe_D = \frac{V_* R_2}{D_1}, \quad Da = \frac{k_f}{R_2^2}, \quad K_{bez} = \frac{K \alpha R_2^2 \rho_1}{\mu_{10}^2}, \quad \bar{D} = \frac{D_2}{D_1}, \quad \Delta = \frac{\delta}{R_2},$$

$$\bar{\rho} = \frac{\rho_2}{\rho_1}, \quad m, \quad \alpha_1 = \frac{\mu_{11}}{\mu_{10}}, \quad \alpha_2 = \frac{\mu_{21}}{\mu_{10}}, \quad \beta = \frac{\mu_{20}}{\mu_{10}}, \quad \bar{\Omega} = \frac{\Omega R_2}{\rho_2 D_1}.$$

Diffusion Peclet number Pe_D includes the velocity $V_* = \mu_{10}/\rho_1 R_2$ and characterizes the relation between convective and diffusion forces; Darcy number Da , relation of elastic and viscous forces K_{bez} and diffusion coefficients \bar{D} together with Pe_D determine the nature of the flow. parameters $\bar{\Omega}$, $\bar{\rho}$, α_1 , α_2 , β are not so significant.

To assess the dimensionless parameters, the following physical values were used that characterize the diffusing fluid (blood) and physical parameters of the capillary [27–31]: $\mu_{10} = 4.5 \times 10^{-3}$ Pa s, $\rho_1 = 1064$ kg/m³, $K = 2.2 \times 10^9$ Pa, $\alpha = 0.3$, $R_1 = 3 \times 10^{-6}$ m, $R_2 = 4 \times 10^{-6}$ m, $D_1 = 2.1 \times 10^{-10}$ m²/s, $k_f = 0.5 \times 10^{-12}$ m². By using data above we can assess the region of alteration of dimensionless complexes: $0.38 \leq Pe_D \leq 1.43 \times 10^3$, $K_{bez} = 5.54 \times 10^4$. In the calculations, the

following dimensionless parameters were varied: m , Pe_D , \bar{D} , Da and p_{10} , p_{20} . The rest of the parameter were fixed: $C_{10} = 0.1$, $C_{20} = 0$, $\xi_1 = 0.75$, $\xi_2 = 1$, $\bar{\rho} = 1$, $\alpha_1 = 0.1$, $\alpha_2 = 0.1$, $\beta = 1$, $\bar{\Omega} = 0$.

The stationary problem for the porous wall (31)–(39) was solved numerically. In differential Eqs. (31), (32) and (34), the convective summand is approximated by the difference against the flow [32]. Such difference provides approximation of the convective summand for any direction of the flow velocity and yields stable algorithm. The initial distribution of velocities and concentrations is specified first. Then the differential equations for concentration and velocity are solved by the double-sweep method. Obtained distributions are used as initial for next iteration. The interface between media is distinguished explicitly. In the direct marching, the coefficients are found with a special approximation of boundary conditions in point at the interface. During reverse marching, we first find C_2 at the interface and then, by using first condition (37) in the point in interface, we find C_1 . The same operations are applied to velocities. The process is repeated until a special condition is met. The calculation is carried out until a special condition is fulfilled—until a solution with a given accuracy is obtained. The variation of spatial steps changes the results no more than 1–5% in the wide region of varying model parameters.

6 Analysis of Results

The Peclet number which characterizes two flow regimes—convective ($Pe_D > 1$) and diffusive one ($Pe_D < 1$)—presents the main interest in the study of fluid transfer through a capillary with porous wall.

At small Peclet numbers, the main contribution to the distribution of concentrations in a capillary is made by diffusion. Since diffusion is a slow process, a smaller amount of a diffusant gets into the porous capillary wall (line 3 in Fig. 5a). With

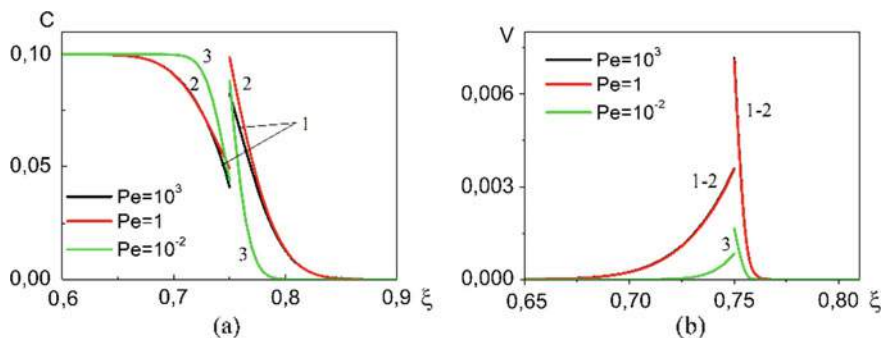


Fig. 5 Distribution of concentration (a) and velocity (b) along the radius at different Peclet numbers 1— $Pe = 10^3$, 2— $Pe = 1$, 3— $Pe = 10^{-2}$, $\bar{D} = 1$, $m = 0.3$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $K_{bez} = 5.54 \times 10^{-2}$, $\Delta = 0.25$, $\bar{\Omega} = 0$

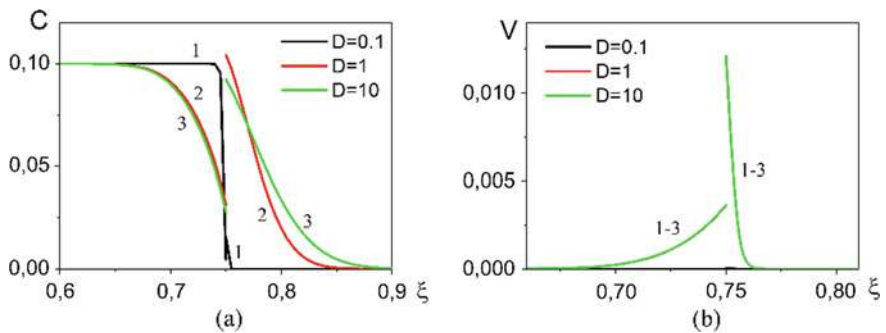


Fig. 6 Distribution of concentration **(a)** and velocity **(b)** along the radius under convective mass transfer at different values of parameter \bar{D} , 1— $\bar{D} = 0.1$, 2— $\bar{D} = 1$, 3— $\bar{D} = 10$, $Pe_D = 10^3$, $m = 0.3$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $K_{bez} = 5.54 \times 10^{-2}$, $\Delta = 0.25$, $\bar{\Omega} = 0$

growing Peclet number ($Pe_D > 1$), the contribution of convective diffusion becomes dominating; the flow velocity is higher in both areas (lines 1 in Fig. 5b). This leads to increased amount of the diffusant permeating the porous wall of the capillary (lines 1 and 2 in Fig. 5a). Line 2 in Fig. 5 corresponds to the transition regime, a convective-diffusive mass transfer. The main changes to concentration and velocity are observed in the vicinity of the interface $\xi = \Delta$.

The redistribution of the diffusant concentration between the materials is appreciably affected by parameter \bar{D} (relation of the diffusion coefficient in area 2 to the diffusion coefficient in area 1). At $\bar{D} > 1$ the fraction of the diffusant in the capillary wall increases (Fig. 6a), while at $\bar{D} < 1$ the diffusant is almost absent in area 2. It has almost zero effect on the velocity distribution (Fig. 6b). The character of concentration distribution both at convective (Fig. 6) and diffusive mass transfer is qualitatively similar under variation of \bar{D} . The difference is only in the velocity values.

Increased capillary wall thickness decreases the concentration and velocity at the interface in both phases both under convective and diffusive mass transfer. This was demonstrated in Table 1, because it was difficult to demonstrate in a figure.

Increased wall porosity decreases the concentration of the diffusant in the second area near the interface; however, the diffusant permeates deeper into the capillary wall under both convective (Fig. 7a, c) and diffusive (Fig. 7b, d) regimes. This is

Table 1 Concentration and velocity at interface for different capillary wall thickness under convective mass transfer

$\Delta = \xi_2 - \xi_1$	$C_1(\xi_1)$	$C_2(\xi_1)$	$V_1(\xi_1)$	$V_2(\xi_1)$
0.25	0.05056	0.16855	0.00368	0.012284
0.3	0.05050	0.16834	0.00367	0.012249
0.35	0.05041	0.16803	0.00366	0.012247
0.4	0.05032	0.16775	0.00367	0.012243
0.45	0.05021	0.16738	0.00367	0.012239

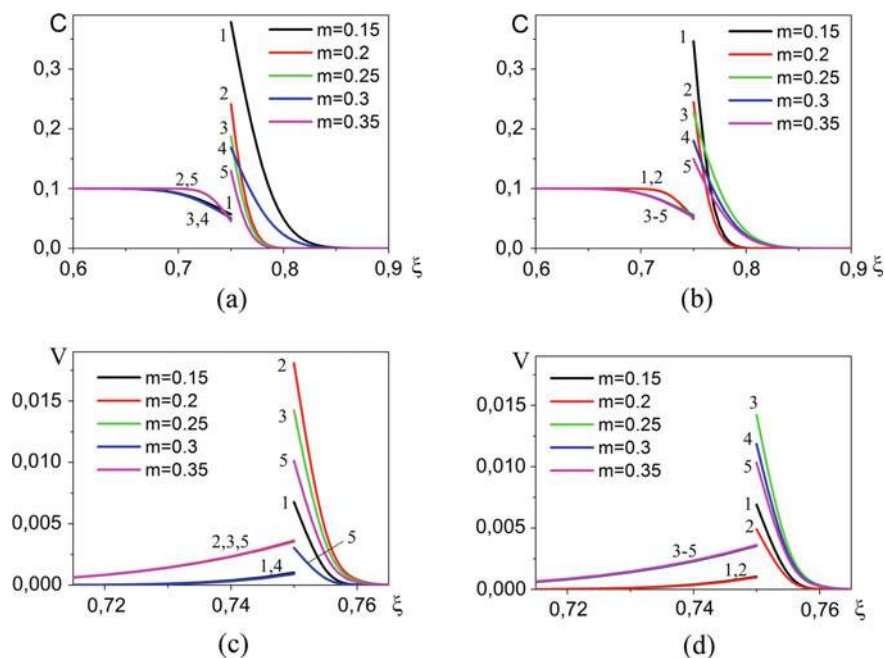


Fig. 7 Distribution of concentration (a, b) and velocity (c, d) along radius under convective (a, c) and diffusive (b, d) mass transfer and different values of parameter m , 1— $m = 0.15$, 2— $m = 0.2$, 3— $m = 0.25$, 4— $m = 0.3$, 5— $m = 0.35$, $\bar{D} = 1$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $K_{bez} = 5.54 \times 10^{-2}$, $\Delta = 0.25$; $\bar{\Omega} = 0$

explained by increased volume of porous space and the diffusant moves more freely in the second area. The variation of porosity hardly affects the concentration distribution in the first area at any Peclet number. The concentration negligibly reduces only near the interface. The velocity behaves ambiguously (Fig. 7c, d) which is due to the interdependence of contrary physical mechanisms.

Decreased permeability (decreased Darcy number) negligibly reduces the concentration and reduces the velocity in both regions at any Peclet number.

The major impact on the velocity distribution is caused by the pressure gradient. An increase in the initial pressure in the first area (not shown) augments the velocity in both areas, while it has almost no effect on the concentration distribution; the concentration drops in both areas only in the interface vicinity. Similar behavior is observed under the diffusive regime of mass transfer. An increase in the initial pressure in the second area has no effect on the concentration distribution, while the velocity in both areas negligibly reduces in any regime.

All previous calculations were made with due regard to the concentration expansion. Parameter α which is included into dimensionless complex $K_{bez} = K \alpha R_2^2 \rho_1 / \mu_{10}^2$. This excites interest in the comparison of the concentration distribution and velocities in the porous wall with and without due consideration of this

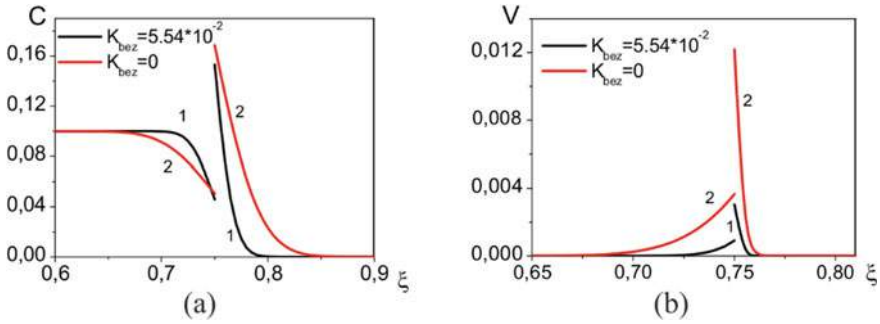


Fig. 8 Distribution of concentration (a) and velocity (b) along the radius under convective mass transfer and for different values K_{bez} , 1— $K_{bez} = 5.54 \times 10^{-2}$, 2— $K_{bez} = 0$, $Pe_D = 10^3$, $\bar{D} = 1$, $m = 0.3$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $\Delta = 0.25$; $\bar{\Omega} = 0$

effect (Fig. 8). Evidently, without consideration of the concentration expansion, the velocity in the interface vicinity increases, which is valid for both sides of the interface. The concentration expansion causes more diffusant to permeate into the porous capillary wall. Such considerable difference is observed even at small values of coefficient K_{bez} .

The effect of viscosity versus concentration on concentration and velocity distributions is illustrated in the Fig. 9. An increase in viscosity with concentration leads to an increase in the fraction of diffusant in the capillary wall in the convective flow regime (Fig. 9a, c), and in the diffusion mode to an insignificant decrease (Fig. 9b, d). In this case, the velocity increases in the convective mode, and decreases in the diffusion mode.

For all figures above it was accepted $\bar{\Omega} = 0$.

The mass flow affects the nature of the concentration distribution in the diffusion mode (Fig. 10b, d red lines), but in the convective mode, no effect is detected (Fig. 10a, c red lines). A smaller amount of diffusant remains in the capillary wall (Fig. 10b) when mass flow is taken into account, the speed also decreases (Fig. 10c).

7 Conclusions

The work suggested a model of biological fluid transfer in a selected macropore with microporous walls with due account for concentration expansion phenomena appearing in stet equation. For two flow regimes—convective ($Pe_D > 1$) and diffusive ($Pe_D < 1$)—we have studied the concentration distribution in the capillary wall. It was shown that the largest impact on the redistribution of concentration between the capillary volume and its porous wall is made by Darcy number Da and correlation of diffusion coefficients. The concentration of the diffusant in the porous layer increases with growing parameter \bar{D} or decreasing porosity or permeability under diffusive mass transfer. The velocity in the interface vicinity increases with rising

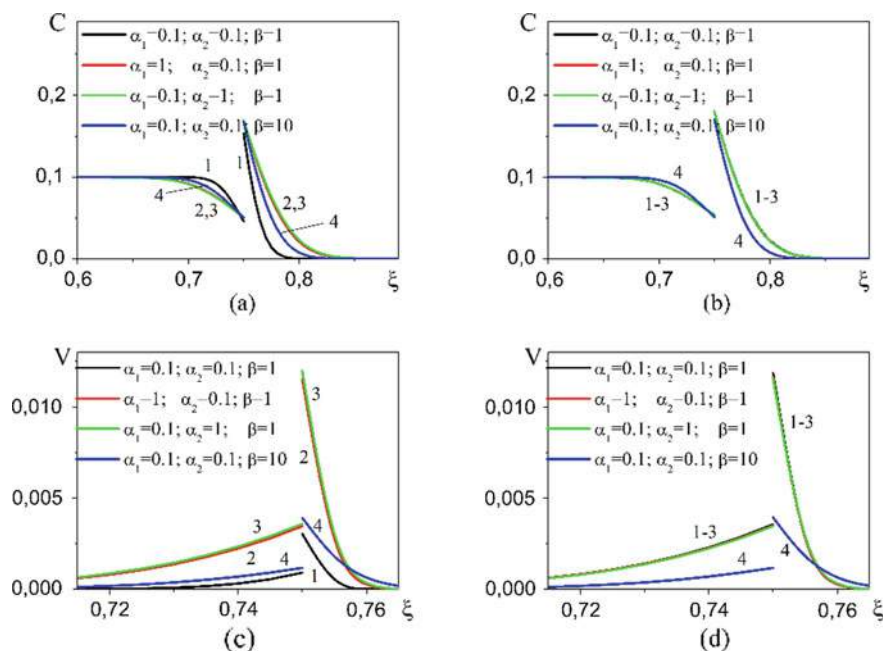


Fig. 9 Distribution of concentration (a, b) and velocity (c, d) along radius under convective (a, c) and diffusive (b, d) mass transfer and different values of viscosity of liquid; $\overline{D} = 1$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $K_{bez} = 5.54 \times 10^{-2}$, $m = 0.3$, $\Delta = 0.25$; $\overline{\Omega} = 0$

pressure in the capillary volume or under decreasing porosity at any Peclet number. It was discovered that the concentration expansion appreciably affects the distribution of velocity and concentration. The ambiguous impact of model parameters on different flow regimes is connected with the interrelation between contrary physical mechanisms. Described model contains practically significant parameters allowing understanding how the concentration distribution changes with flow type variation.

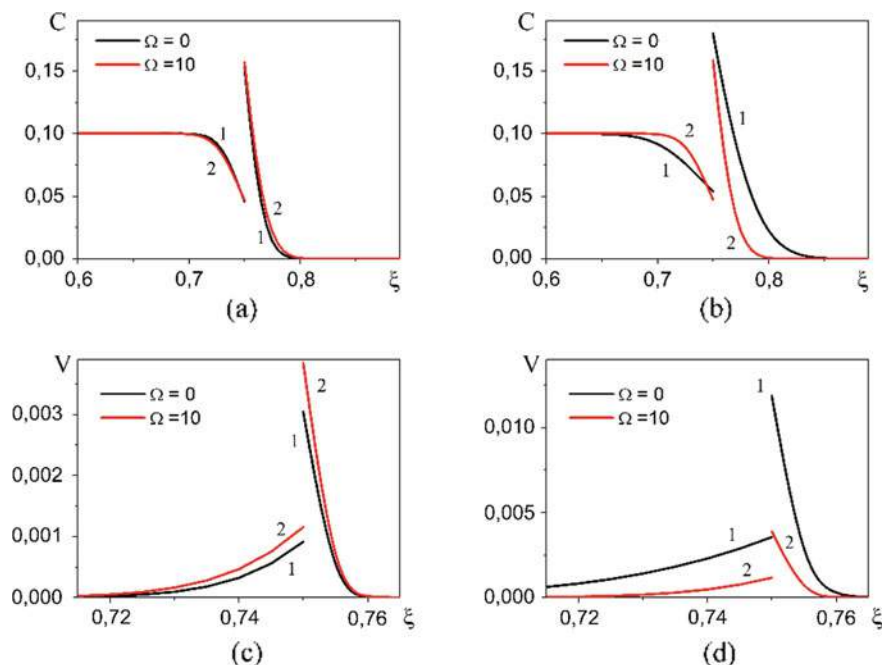


Fig. 10 Distribution of concentration (a, b) and velocity (c, d) along radius under convective (a, c) and diffusive (b, d) mass transfer and different values of parameter $\bar{\Omega}$, 1— $\bar{\Omega} = 0$, 2— $\bar{\Omega} = 10$, $\bar{D} = 1$, $Da = 0.01$, $p_{10} = 2$, $p_{20} = 0.2$, $K_{bez} = 5.54 \cdot 10^{-2}$, $m = 0.3$, $\Delta = 0.25$

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Failure Mechanisms of Alloys with a Bimodal Grain Size Distribution



Vladimir A. Skripnyak, Evgeniya G. Skripnyak, and Vladimir V. Skripnyak

Abstract A multi-scale computational approach was used for the investigation of a high strain rate deformation and fracture of magnesium and titanium alloys with a bimodal distribution of grain sizes under dynamic loading. The processes of inelastic deformation and damage of titanium alloys were investigated at the mesoscale level by the numerical simulation method. It was shown that localization of plastic deformation under tension at high strain rates depends on grain size distribution. The critical fracture stress of alloys depends on relative volumes of coarse grains in representative volume. Microcracks nucleation at quasi-static and dynamic loading is associated with strain localization in ultra-fine grained partial volumes. Microcracks arise in the vicinity of coarse and ultrafine grains boundaries. It is revealed that the occurrence of a bimodal grain size distributions causes increased ductility, but decreased tensile strength of UFG alloys. The increase in fine precipitation concentration results not only strengthening but also an increase in ductility of UFG alloys with bimodal grain size distribution.

Keywords Bimodal distribution of grain sizes · Computational plasticity · High strain rates · Titanium alloys · Magnesium alloys

1 Introduction

Hexagonal close-packed (HCP) metals, such as Ti, Mg, Zn and Zr, are of interest for engineering and medical applications due to their unique combination of high ductility and strength. In recent years, intensive efforts have been focused on investigations of the physical and mechanical properties of alloys, whose grain structure is formed by surface rolling, severe plastic deformation, selective laser melting, or selective laser sintering, and friction stir processing [1–5].

Recently Long et al. showed a bimodal microstructure formation in the ultrafine-grained Ti–6Al–4V alloy produced by means of the spark plasma sintering of

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a mixture of ball-milled and unmilled powders [6]. The additive manufacturing technology of HCP metals opens new possibilities for obtaining parts from alloys combining high strength, ductility, and fracture toughness.

Usually, the formation of nano-sized and ultra-fine grained structures in metal alloys leads to increased strength and decreased ductility. The low ductility of ultra-fine (UFG) alloys limits their applications as advanced engineering materials. In recent years it was found that the alloys with a bimodal grain size distribution may exhibit increased yield strength without reducing ductility [7, 8]. Guo showed the ductility of fine-grained (FG) Zr with an average grain size of 2–3 μm was 17% higher than that of coarse-grained alloy [2]. Pozdnyakov [9], Malygin [10], Ulacia [11] showed that the bimodal grain distribution can be formed in light alloys by means of severe plastic deformation and follow-up heat treatment. It was revealed that hexagonal close-packed (HCP) alloys with a bimodal grain size distribution exhibit a number of anomalies in mechanical behavior [12, 13]. Therefore, the optimization of strength and ductility requires a profound knowledge of volume fraction and distribution of ultra-fine grains and coarse grain in multi-modal alloys. Research of the influence of the grain structure on the strength and ductility of the alloys under cyclic loadings, dynamic loadings, quasistatic loadings are carried out using experimental methods and by computer simulation [14–17].

Berbenni proposed a theoretical micromechanical model taking into account the grain size distribution and representing the accommodation of grain deformation in HCP metals (Zirconium- α) [7]. Raeisinia proposed a model to examine the effect of unimodal and bimodal grain size distributions on the uniaxial tensile behavior of a number of polycrystals [12]. The model demonstrated that bimodal grain size distributions have enhanced macro mechanical properties as compared with their unimodal counterparts. The authors of this article proposed a multiscale model of UFG metals with bimodal grain size to predict the localization of plastic flow in UFG light alloys under dynamic loads with respect to the ratio of volume concentrations of small and large grains. Results of calculations have shown increased dynamic ductility of UFG titanium and magnesium alloys, when the specific volume of coarse grains is greater than 30%.

Zhu proposed a micromechanics-based model to investigate the mechanical behavior of polycrystalline dual-phase metals with a bimodal grain size distribution, and fracture by means of nano/microcracks generation during plastic deformation. Results have shown that the volume fraction of coarse grains controls the strength and ductility of metals [18–21]. Clayton developed a continuum model of crystal plasticity for the interpretation of experimental data on shock wave propagation and spall fracture in polycrystalline aggregates [22]. The model accounted for complex features of the mechanical response of alloys: plastic anisotropy, large volumetric strain, heat conduction, thermos-elastic heating, rate- and temperature-dependent flow stress. Magee proposed a multiscale modeling technique to simulate the microstructural deformation of an alloy with bimodal grain size distribution. The simulation has shown that intergranular cracks nucleate at the coarse grains/ultra-fine grains interfaces [23]. Although much research is carried out on the mechanical properties of alloys with a bimodal grain size distribution, a number of issues remains

poorly understood. The effect of the bimodal grain structure of alloys on the ultimate strain to failure at high strain rates is poorly studied [24]. The laws of localization of plastic deformation and the laws of damage accumulation in HCP alloys with a bimodal grain size distribution under dynamic impacts are not well studied. These problems are of great practical importance and are associated with the transition to new technologies for digital design and production of technical and medical products.

In this paper, we develop a multilevel approach to study the mechanical behavior of HCP alloys in a wide range of strain rates.

2 Computational Model

Titanium, zirconium, and magnesium alloys with a hexagonal closed-packed (HCP) structure have a significant low crystal symmetry compared to the face-centered cubic (FCC) and body-centered cubic (BCC) crystal structures of engineering alloys. The mechanical behavior of HCP alloys under quasistatic and dynamic loading at temperatures T/T_m less than 0.5 is determined by dislocation mechanisms and twinning [25–29]. Authors used modifications of the constitutive equations developed in the framework of the micro-dynamical approach and taking into account the thermally activated dislocation mechanisms [15–17, 30–32].

The grain sizes and the grain-boundary phase structure affect the glide of dislocations and the formation of dislocation substructures during the plastic flow [27].

Mechanical response of polycrystalline alloys can be described by parameters of states averaging over the model representative volume element (RVE). Therefore, it is required for model volume to represent not only a given grain size distribution but also realistic values of the physical properties of the system.

We use the approach of a finite mixture model for the estimation of average sizes in multimodal grain size distribution [7, 17, 33].

The multimodal grain size distribution described by probability density function $g(d_g)$. Distribution $g(d_g)$ is a mixture of k component distributions g_1, g_2, g_3 of ultra-fine grains (UFG), fine grains (FG), coarse grains (CG), respectively:

$$g(d_g) = \sum_{k=1}^m \lambda_k g_k(d_g), \quad (1)$$

where λ_k are the mixing weights, $\lambda_k > 0$, $\sum_{k=1}^3 \lambda_k = 1$, $k = 1, 2, 3$.

This method allows us to determine the multimodal grain size distribution function and the range of grain size groups (UFG, FG, CG). In this paper, we present simulation results for unimodal and bimodal (UFG and CG) grain distribution. We used experimental data of grain size distribution of alloys after severe plastic deformation to calibrate the computational model.

The RVE can be created using the experimental data on grain structures obtained by the analysis of Electron Backscatter Diffraction (EBSD) based on scanning electron microscopy (SEM) [34–36]. This analysis gives quantitative information about sizes and shapes of grains, Euler angles, angles of misorientation at the grain/subgrain boundaries with angular resolution $\sim 0.5^\circ$. The shape coefficient was determined by the relation of minimal grain size to maximal size $\left(\xi_k = d_g^{\min}/d_g^{\max}\right)_k$, where k is the number of grain size group.

The analysis of grain structure distributions has shown that there are several types of grains structure characterized by unimodal (near log-normal) distribution, and bimodal grain size distribution, multimodal grain size distribution [2, 7, 8, 35].

The specific volume of ultrafine grains (with grain size $100 \text{ nm} < d_g < 1 \mu\text{m}$), microstructural grains ($1 < d_g < 10 \mu\text{m}$), and coarse grains ($10 \mu\text{m} < d_g < d_g^{\max}$) was estimated using the probability density function $g(d_g)$ of grain size distribution:

$$C_{UFG} = \int_{d_g^{\min}}^{1 \mu\text{m}} g_1(x) dx, \quad C_{FG} = \int_{1 \mu\text{m}}^{10 \mu\text{m}} g_2(x) dx, \quad C_{CG} = \int_{10 \mu\text{m}}^{d_g^{\max}} g_3(x) dx, \quad (2)$$

where C_{UFG} , C_{FG} , C_{CG} are specific volumes of UFG, FG, and CG, g_1 , g_2 , g_3 are the probability density functions of UFG, FG, and CG systems, respectively.

The relative volume of coarse and ultra-fine grains in model RVE was determined in accordance with the probability density function of the grain size distribution.

Figure 1 shows a 3D RVE model of a titanium alloy with a bimodal grain size distribution. An RVE with dimensions of $14 \times 8 \times 1 (\mu\text{m})^3$ was used.

Computational domains were meshed with eight-node linear bricks and reduced integration together with hourglass control.

The kinematic boundary conditions correspond to macroscopic tension. The scheme of boundary conditions is shown in Fig. 1.

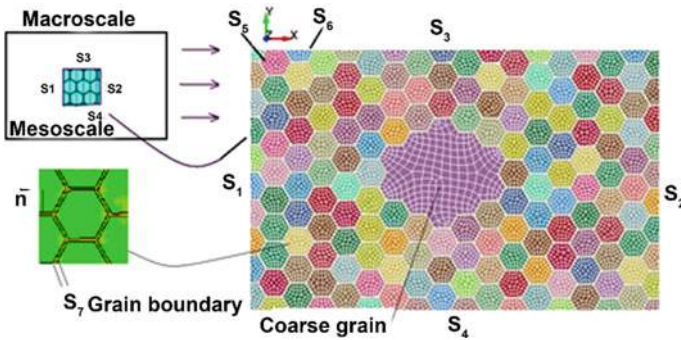


Fig. 1 Scheme of boundary conditions

$$\begin{aligned}
u_x(x_k, t) &= 0, \quad x_k \in S_1 \\
u_x(x_k, t) &= v_x, \quad x_k \in S_2 \\
\sigma_{22} &= 0, \quad x_k \in S_3, \quad x_k \in S_4 \\
u_z &= 0, \quad x_k \in S_5, \quad x_k \in S_6 \\
u_k^A - u_k^B &= 0, \quad \sigma_n^A = -\sigma_n^B, \quad k = 1, 2, 3, \quad x_k \in S_7
\end{aligned} \tag{3}$$

where u_k^A, u_k^B are the projections of the displacement rate onto the external normal to the boundary S_7 of the grain and the grain boundary phase at the boundary points, respectively, σ_n^A, σ_n^B are the components of the surface forces along the external normal \vec{n} to the boundary S_7 , respectively, x_k is the Cartesian coordinate.

Mechanical behavior at the mesoscale level is described within the approach of the damaged elastic-plastic medium. The system of equations includes:

Conservation equations (4);

Kinematic relations (5);

Constitutive relations (8);

Equation of State (9);

Relaxation equation for the deviatoric stress tensor (10).

$$\frac{d\rho}{dt} = \rho \frac{\partial u_i}{\partial x_i}, \quad \frac{\partial \sigma_{ij}}{\partial x_j} = \rho \frac{du_i}{dt}, \quad \rho \frac{dE}{dt} = \sigma_{ij} \dot{\varepsilon}_{ij}, \tag{4}$$

where ρ is the mass density, u_i is the components of the particle velocity vector, x_i is the Cartesian coordinates, $i = 1, 2, 3$, E is the specific internal energy, σ_{ij} are the components of the effective stress tensor of the damaged medium, $\dot{\varepsilon}_{ij}$ are the components of strain rate tensor.

Kinematics of the medium was described by the local strain rate tensor:

$$\dot{\varepsilon}_{ij} = (1/2)[\partial u_i / \partial x_j + \partial u_j / \partial x_i], \quad \dot{\omega}_{ij} = (1/2)[\partial u_i / \partial x_j - \partial u_j / \partial x_i], \tag{5}$$

where $\dot{\varepsilon}_{ij}, \dot{\omega}_{ij}$ are the components of strain rate tensor and the bending-torsion tensor, u_i is the component of particles velocity vector.

The components of strain rate tensor are expressed by the sum of elastic and inelastic terms:

$$\dot{\varepsilon}_{ij} = \dot{\varepsilon}_{ij}^e + \dot{\varepsilon}_{ij}^p, \quad \dot{\varepsilon}_{ij}^p = \dot{\varepsilon}_{ij}^p + \delta_{ij} \dot{\varepsilon}_{kk}^p / 3, \tag{6}$$

where $\dot{\varepsilon}_{ij}^e$ are the components of the elastic strain rate tensor, $\dot{\varepsilon}_{ij}^p$ are the components of the inelastic strain rate tensor.

The bulk inelastic strain rate is described by relation:

$$\dot{\varepsilon}_{kk}^p = \dot{f}_{growth} / (1 - f), \tag{7}$$

where f is the damage parameter, the substantial time derivative is denoted via dot notation.

The bulk inelastic strain rate ε_{kk}^p is equal to zero only when the material is undamaged.

$$\sigma_{ij} = \sigma_{ij}^{(m)} \varphi(f), \quad \sigma_{ij}^{(m)} = -p^{(m)} \delta_{ij} + S_{ij}^{(m)}, \quad (8)$$

$$\begin{aligned} p^{(m)} &= p_x^{(m)}(\rho) + \Gamma(\rho) \rho E_T, \quad E_T = C_p T, \\ p_x^{(m)} &= \frac{3}{2} B_0 \cdot ((\rho_0/\rho)^{-7/3} - (\rho_0/\rho)^{-5/3}) \left[1 - \frac{3}{4} (4 - B_1) \cdot ((\rho_0/\rho)^{-2/3} - 1) \right] \end{aligned} \quad (9)$$

$$DS_{ij}^{(m)}/Dt = 2\mu(\dot{\varepsilon}_{ij}^e - \delta_{ij}\dot{\varepsilon}_{kk}^e/3), \quad \dot{\varepsilon}_{ij}^p = \lambda \partial \Phi / \partial \sigma_{ij}, \quad (10)$$

the function $\varphi(f)$ establishes a relation between the effective stresses of the damaged medium and the stresses in the condensed phase, Γ is the Grüneisen coefficient, ρ_0 is the initial mass density of the condensed phase of the alloy, γ_R , ρ_R , n , B_0 , B_1 are the material's constants, C_p is the specific heat capacity, $D(\cdot)/Dt$ is the Jaumann derivative, μ is the shear modulus, \dot{f}_{growth} is the void growth rate, f is the void volume fraction in the damaged medium, $\dot{\lambda}$ is the plastic multiplier derived from the consistency condition $\dot{\Phi} = 0$, and Φ is the plastic potential. The plastic potential was described using the Gurson–Tvergaard model (GTN) [32, 37–39].

The Grüneisen coefficient Γ was equal to 1.42 and 1.09 for Mg–3Al–1Zn and Ti–5Al–2.5Sn, respectively.

The function $\varphi(f)$ takes the form of $(1 - f)$ for pressure and is implicitly defined for the deviatoric stress tensor [40].

The temperature rise associated with energy dissipation during plastic flow can be evaluated by relation [25, 32]:

$$T = T_0 + \int_0^{\varepsilon_{eq}^p} (\beta/\rho C_p) \sigma_{eq} d\varepsilon_{eq}^p, \quad (11)$$

where T_0 is the initial temperature and $\beta \sim 0.9$ is the parameter representing a fraction of plastic work converted into heat.

The specific heat capacity for Ti–5Al–2.5Sn titanium was calculated by the phenomenological relations within the temperature range 293–1115 K [32]:

$$C_p = 248.389 + 1.53067T - 0.00245T^2 \text{ (J/kg K) for } 0 < T < T_{\alpha\beta} = 1320 \text{ K}, \quad (12)$$

The temperature dependence of the shear modulus for alpha titanium alloy was described by the equation:

$$\mu(T) = 48.66 - 0.03223T \text{ (GPa)}, \quad (273\text{ K} < T < 1200\text{ K}). \quad (13)$$

The flow stress was described by equation:

$$\sigma_s = \sigma_{s0} \exp\left\{C_1 \sqrt{(1 - T/T_m)}\right\} + C_2 \sqrt{1 - \exp\{-k_0 \varepsilon_{eq}^p\}} \exp\{-C_3 T\} \exp\{C_4 T \ln(\dot{\varepsilon}_{eq}/\dot{\varepsilon}_{eq0})\}, \quad (14)$$

where $\dot{\varepsilon}_{eq} = [(2/3)\dot{\varepsilon}_{ij}\dot{\varepsilon}_{ij}]^{1/2}$, $\dot{\varepsilon}_{eq0} = \gamma_1 \exp\{-T/\gamma_2\} + \gamma_3$, $\varepsilon_{eq}^p = \int_0^t \dot{\varepsilon}_{eq}^p dt$ is the equivalent plastic strain $\gamma_1 = 2115.08615 \text{ s}^{-1}$, $\gamma_2 = 38.26589 \text{ K}$, $\gamma_3 = 9.82388 \times 10^{-5} \text{ s}^{-1}$ and T_m is the melting temperature, $\dot{\varepsilon}_{eq0} = 1.0 \text{ s}^{-1}$.

$$\sigma_s = \sigma_{s0} + C_5(\varepsilon_{eq}^p)^{n_1} + k_{hp}d_g^{-1/2} - C_2 \exp\{-C_3 T + C_4 T \ln(\dot{\varepsilon}_{eq}/\dot{\varepsilon}_{eq0})\}, \quad (15)$$

where σ_{s0} , C_5 , n_1 , k_{hp} , C_2 , C_3 , C_4 are the material parameters, d_g is the average grain size.

The impact of grain size distribution on the stress of hcp alloys with average grain sizes in single-mode and bimodal microstructures was taken into account in Eq. (15) by analogy with the Hall-Petch relation.

Material parameters of Ti-5Al-2.5Sn and Mg-3Al-1Zn are shown in Tables 1 and 2, respectively.

The influence of damage on the flow stress was taken into account using the Gurson-Tvergaard model [32, 36, 38]:

$$(\sigma_{eq}^2/\sigma_s^2) + 2q_1 f^* \cosh(-q_2 p/2\sigma_s) - 1 - q_3 (f^*)^2 = 0, \quad (16)$$

where σ_s is the yield stress and q_1 , q_2 and q_3 are the model parameters, $\sigma_{eq} = \sqrt{\frac{3}{2}\sigma_{ij}\sigma_{ij} - \frac{1}{2}\sigma_{kk}\sigma_{kk}}$.

$$\begin{aligned} \dot{f} &= \dot{f}_{nucl} + \dot{f}_{growth}, \\ \dot{f}_{nucl} &= \varepsilon_{eq}^p (f_N/s_N) \exp\{-0.5[(\varepsilon_{eq}^p - \varepsilon_N)/s_N]^2\}, \\ \dot{f}_{growth} &= (1 - f)\dot{\varepsilon}_{kk}^p, \end{aligned} \quad (17)$$

Table 1 Material parameters of Eq. (14)

Parameter	σ_{s0} (GPa)	C_1	C_2 (GPa)	C_3 (K ⁻¹)	C_4 (K ⁻¹)	k_0	T_m (K)
Ti-5Al-2.5Sn	0.02	3.85	0.56	0.0016	0.00009	8.5	1875

Table 2 Material parameters of Eq. (15)

Coefficients	σ_{s0} (GPa)	k_{hp} (Pa nm ^{1/2})	C_2 (GPa)	C_3 (K ⁻¹)	C_4 (K ⁻¹)	C_5 (GPa)	n_1
Mg-3Al-1Zn	0.141	6.2	0.315	0.0029	0.000389	0.505	0.2514

Table 3 Dimensionless parameters for the Gurson–Tvergaard–Needleman (GTN) model for alpha titanium and magnesium alloys

	q_1	q_2	q_3	f_0	f_N	f_c	f_f	ε_N	S_N
Ti–5Al–2.5Sn	1.3	1	1.69	0.00	0.2	0.035	0.4	0.28	0.1
Mg–3Al–1Zn	1.5	1	2.25	0.00	0.4	0.073	0.128	0.1	0.2

where ε_N and S_N are the average nucleation strain and the standard deviation, respectively. The amount of nucleating voids is controlled by the parameter f_N

$$\begin{aligned} f^* &= f \text{ for } f \leq f_c; \\ f^* &= f_c + (\bar{f}_F - f_c)/(f_F - f_c) \text{ for } f > f_c, \end{aligned} \tag{18}$$

where $\bar{f}_F = (q_1 + \sqrt{q_1^2 - q_3})/q_3$.

The final stage in ductile fracture comprises the voids coalescence [38]. This causes softening of the material and accelerated growth rate of the void fraction f^* .

The model parameters for Ti–5Al–2.5Sn and Mg–3Al–1Zn were determined using numerical simulation. These parameters are given in Table 3.

We use the ductile fracture criteria for alloys at the room and elevated temperatures owing to relatively low melting temperature [31, 38, 41].

Finite elements are removed from the grid model and free boundary conditions are introduced at the formed boundary, when the local fracture criterion is met.

Specific features of mechanical properties of nanostructured materials are connected to distinctions of matter properties in a crystalline phase of grains and in the boundary of grains. A decrease in the mass density of nanostructured materials is caused by increased defect’s density and relative volume of grain boundary phase.

3 Results and Discussion

Figure 2 shows the fields of equivalent plastic strain under tension at $v_x = 2.3$ m/s. Damages were localized near the grain boundary of coarse grains.

Damage nucleation in alloys with a bimodal grain size distribution occurs in shear bands and zones of their intersection.

These results agree with experimental observations of strain localization and fracture of titanium alloys by Sharkeev [42], authors of this work [32, 43], Valoppi [44] and Zheng [45]. It is significant that increases in fine precipitates concentration in alloys caused the increase in resistance to plastic flow within both coarse and ultrafine grains.

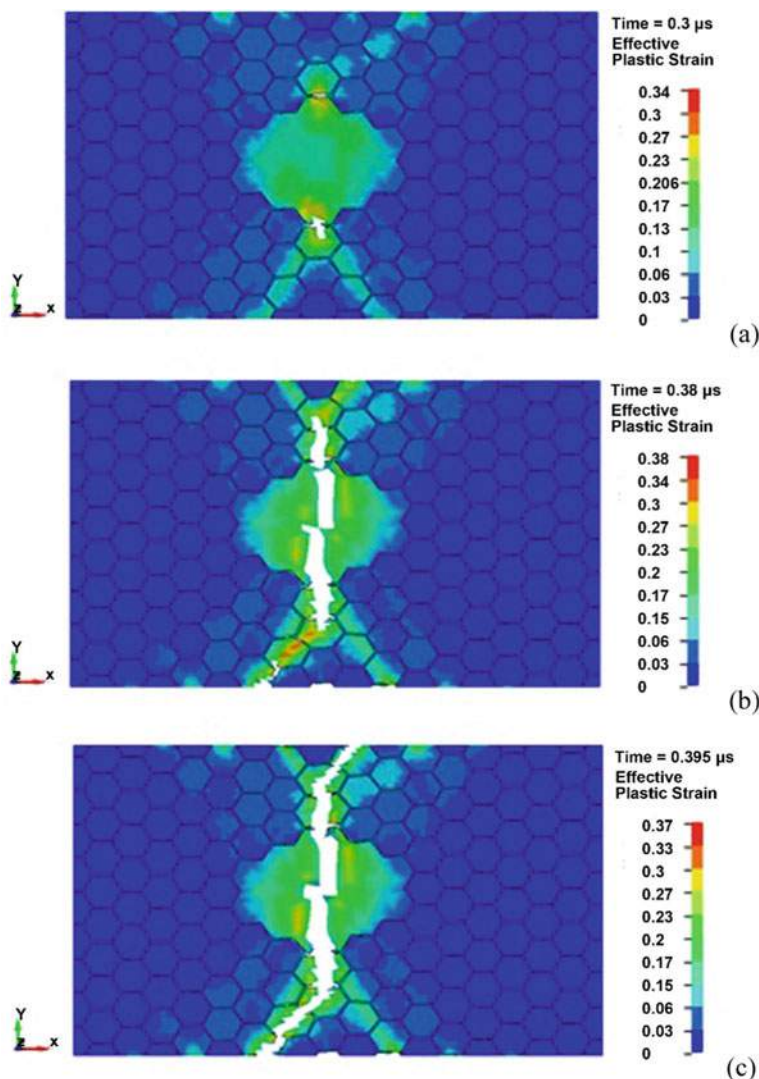


Fig. 2 Effective plastic strain field at time, **a** 0.332 μs , **b** 0.3895 μs , **c** 0.398 μs in Ti–5Al–2.5Sn alloy with bimodal grain structure

Thus, the fracture of alloys with bimodal grain structures is caused by the damage nucleation at the boundary of coarse grains with an ultrafine-grained structure and further growth of damage in mesoscopic bands of localized plastic deformation.

Thus, the process of fracture of alloys with bimodal grain structures is associated with the formation of mesoscopic bands of localized plastic deformation.

The segregation of impurity atoms in the grain-boundary phase affects the formation of plastic shear bands [14, 46].

The mechanical properties of the grain boundary phase were varied to take into account the effect of segregation of impurity atoms on the yield stress in the simulation.

Figure 2b shows the field of equivalent strains, indicating that the formation of cracks in large grains can be accompanied by a change in the orientation of the shear localization bands at the mesoscopic level. This occurs as a result of a change in the orientation of the plane of maximum shear stresses during the evolution of a triaxial stress state near cracks. A macroscopic crack obtained by modeling the tension of flat samples has the same configuration as the cracks observed in experimental studies by Verleysen [24] and authors of this work [31, 32].

The averaged strain along the axis of tension of the computational domain at the moment of crack crossed the representative volume was interpreted as the ultimate strain to fracture of the alloy.

Figure 3 shows the dependence of strain to fracture of titanium and magnesium alloys on the logarithm of strain rates. Experimental data reported by Ulacia for coarse-grained Mg–3Al–1Zn alloy are marked by filled square symbols [11]. The experimental data for the Ti–5Al–2.5Sn alloy are shown by filled triangular symbols [32, 47].

Strains to fracture behave nonmonotonically and nonlinearly with increasing strain rate in the range from 0.001 to 1000 1/s as shown in Fig. 3 (Curves 1 and 3). In coarse-grained HCP alloys, this is related to more intense twinning under dynamic loading.

Curve 2 is obtained by the approximation of calculated values of the strains to fracture under tension of a magnesium alloy with a bimodal grain structure with a specific volume of large grains of 70%. With an increase in the concentration of micron and

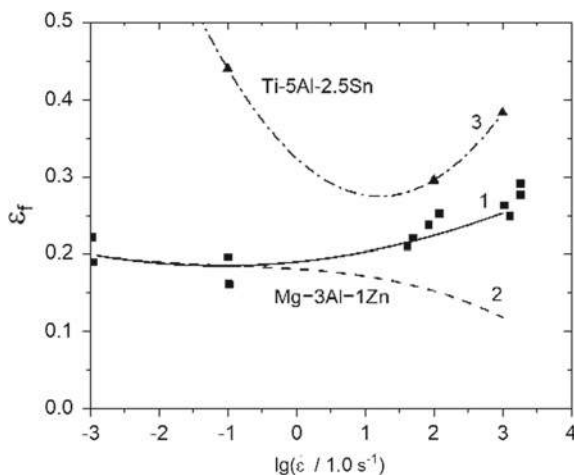


Fig. 3 The strain to fracture versus logarithm of strain rates for magnesium and titanium alloys with a unimodal (curve 2, 3) and a bimodal grain sizes distribution (curve 1, the concentration of coarse grains is 70%)

submicron grains in the alloy, the strain to failure under tension decreases nonlinearly. The simulation results agree with the available experimental data [11, 45–50].

The strain to fracture of alloy with bimodal grain size distribution versus specific volume of coarse grains under quasi-static tension can be described by the relation:

$$\varepsilon_f^n = 0.01 \exp(C_{cg}/0.363), \quad (19)$$

where ε_f^n is the strain to fracture under quasi-static tension, C_{cg} is the specific volume of coarse grains.

Equation (19) describes the ductility of alloys with a bimodal grains distribution versus the specific volume of coarse grains. The increase in ductility of HCP alloys under quasi-static tension occurs when a specific volume of coarse grains is greater than 30%.

4 Conclusions

The multiscale approach was used in the computer simulation of fracture of magnesium and titanium alloys at high strain rates. Structured RVEs were proposed to predict the mechanical properties of alloys taking into account the grain size distribution and the segregation impurity atoms in the grain-boundary.

The results of computer simulation showed that damage nucleation in alloys with a bimodal grain size distribution occurs in the shear bands and their intersection zones. Damage arises at the boundary between coarse- and ultrafine-grained structures. Further damage growth occurs in the mesoscopic bands of localized plastic deformation.

Thus the computer simulation can be used to estimate the influence of grains size distribution on the dynamic strength and ductility of HCP alloys.

Localization of plastic flow in HCP alloys with bimodal grain size distribution under tension at high strain rates depends on the ratio between volume concentrations of fine and coarse grains. As a result, the strain to fracture of hcp alloys with bimodal grain size distribution varies nonlinearly with tensile strain rate in the range from 0.001 to 1000 1/s.

The dynamic ductility of HCP alloys with bimodal grain size distribution is increased when a specific volume of coarse grains is greater than 30%.

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Self-reproduction Cycles of Living Matter and Energetics of Human Activity



Leonid E. Popov

Abstract In the author's opinion, many global problems that face humanity—in the fields of education, medicine, management etc. can be tackled more effectively if the cyclic nature of self-reproducing systems—including living beings—is taken into account. Summarizing the main physiological findings of the last decades on “adaptation reactions”, one can very roughly say that the way of action which is *effective* in the sense of productive activity of people happens at the same time to be *healthy*, and it gives the participants of the process the feeling of *happiness*. The present paper represents a very short overview of the contemporary concepts of the adaptation reactions based on the fundamental understanding of their cyclic nature due to general properties of self-reproducing systems. One interesting feature of self-reproduction cycles is its first “phase of orientation” which was not discussed in detail in the past but plays a key role in the whole cycle.

Keywords Self-reproducing systems · Life · Adaptation reactions · Activation · Training · Stress · Health · Memory · Catabolic phase · Anabolic phase

1 Introduction

Any activity of living systems is cyclic at all levels of the organization of life. The cyclic character of activity of biological systems is associated not only with the periodic nature of the action of external factors on it (such as the rotation of the Earth around its axis, its movement around the Sun or the cyclic change in solar activity). The above mentioned cycles are the result of adaptation of living systems to periodic changes in the environment. However, life is cyclic in itself, by virtue of its very essence. Let us first briefly discuss this fundamental aspect of life which is essential for the following discussion.

How does “life” differ from inert matter? This question was put forward many times in the history of science. In retrospect, we can state that there is one point where

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many researchers do agree: the essential property of living systems is their “reproduction”. Let us consider this property closer. Chilean neuroscientists H. Maturana and F. Varela provide the following explanation: “When we speak of living beings, we presuppose something in common between them; otherwise we wouldn’t put them in the same class we designate with the name ‘living’. ... Our proposition is that living beings are characterized in that, literally, they are continually self-producing. We indicate this process when we call the organization that defines them an *autopoietic organization*.”

What is distinctive about... <living beings> ... is that their organization is such that their only product is themselves, with no separation between producer and product. The being and doing of an autopoietic unity are inseparable, and this is their specific mode of organization” [1].

So living matter only exists due to continuously reproducing itself. The word “self-reproduction” very precisely reflects the meaning of the basic characteristic of all life: for all interactions with the environment, the living beings remain “themselves”, although in the process of interaction parameters of a living system can significantly deviate from their values at steady state. The deviation is followed by a return to the stationary state of “stable non-equilibrium”.

Self-reproduction means turning back to oneself. It would seem that there is no other interpretation here. But in the popular, as well as in scientific literature, self-reproduction is very often considered a synonym for reproduction of generations, and when speaking of “self-reproduction cycles”, one usually means a cycle of producing a new generation. However, this aspect means the reproduction of population and the biological species as a whole, while each living unit reproduces itself continuously during its whole life. In the present paper, we focus on this—more unusual, but in reality central aspect of living things—their continuous self-reproduction during the course of individual life.

The process of individual self-reproduction is also cyclical in nature: after deviating from a stationary state, the system returns to the same state. Ugolev [2] formulated the principle of cyclic activity among the “principles of natural technologies of biological systems”: “At all levels of organization (from cellular to planetary), biological systems (more precisely, processes in them) are partially or completely cyclized... The principle of recurrence is one of the most important principles to ensure maximum efficiency and effectiveness of living systems through the multiple use of the same structures. Cyclization also ensures coordination of all components, implementing a multi-stage process. Many process systems considered by us as linear will be later characterized as cyclic...”.

2 Three Phases of the Self-reproduction Cycle

Thus, our life activity is discrete, and each cycle proceeds in phases. Let us first analyze the general structure of each such a cycle. Each behavioral act begins with the appearance in our mind of an *image of the result of our action* [3]. The formation

of this image is an elementary event of prognostic activity of the nervous system. We can talk about the prognostic phase of the cycle. The two subsequent steps are obvious. The second phase is the *action* itself, leading to the achievement of the result using the functional system that was formed at the first stage. During this phase, the work is performed to achieve the desired (predicted) result, and this work is associated with the expenditure of free energy of the body. Finally, the third phase is the *return of the organism to the steady state of stable disequilibrium*. During this phase, the body recovers the energy and possibly structural costs related to the work done in the second phase.

We note two points here. Firstly, the formation of an image of the result of an organism's action is associated with the extraction of some information stored in long-term memory in the central nervous system. In addition, the formation of a functional system corresponding to this image requires preparation for the action of executive bodies. That is, the entire first phase of the cycle is associated with the processes of structure formation, with the processes of *anabolism*.

In the second phase, work is performed that is accompanied by the dissipation of free energy of the body, that is, with the cost of reserves of energy and, possibly, structure. Thus, the second phase is a *catabolic* one. The cost of energy and body structure is normally restored *excessively* in the third phase. Moreover, the processes of anabolism occur only in the structures that took part in the action.

Phenomenologically, the existence of the prognostic phase, during which the action is still absent, was of course known for a very long time, in various contexts, for example under the name of "keeping a clear head". "Keeping a clear head" in an acute situation does not mean not acting at all. "Clear head" refers first of all to the initial phase of orientation and forecast, without which the active action turns into "action nowhere." The absence of a phase of the formation of the image of the result of the action would lead to a chaotic, senseless and potentially dangerous activity for the body, the "panic". The physiological reaction characteristics of this first phase of the behavioral cycle have also been repeatedly described: The heart first "stops" (that is, the pulse drops sharply), and only then begins to "beat wildly". This physiological reaction can be found in such idiomatic expressions as a "heart-stopping event" (something that is very impressive or exciting).

However, it was this first phase of the cycle that was last to be discovered and described in the framework of physiological studies. A single cycle of physiological self-reproduction was studied by Arshavsky with collaborators in the laboratory of age-related physiology and pathology of animals and humans at the Institute of Experimental Medicine since 1935. By this time, researchers have already come to the understanding that living time is discrete (although it is perceived as flowing continuously): it is measured by metabolic cycles, which, as it was originally thought, consist of two phases: catabolic and anabolic. In the catabolic phase, motor activity is carried out, and work is done due to the energy of the destruction of cellular substrates. In the anabolic phase, the disturbed structure is restored, and recovery occurs to an extent exceeding the costs in the second phase. This transfers the body to a higher than the initial level of disequilibrium and ordering.

Bearing in mind this excessive anabolism, I. A. Arshavsky called the metabolic cycle associated with a single motor activity a hypercycle.

But only in 1968 Arshavsky experimentally established the presence of another phase of the cycle: the “phase of the starting metabolism”. Studies were conducted on the human fetus: through the wall of the abdomen of the mother, the fetal heartbeat was registered. It was expected that with recurrent “spontaneous” increase in activity would be associated increasing of the heart rate (catabolic processes occur with the dominance of the sympathetic nervous system that increases the heart rate), and then of slowing down (anabolism, dominance of the parasympathetic nervous system). It was found, however, that before each motor activity, the heartbeat slowed down (parasympathetic reaction, anabolism). After that, extensive research on puppies was undertaken during the process of prenatal development. On the obtained cardiograms, you can see all three phases associated with motor activity: a few rarer pulses (the starting anabolism phase), a rapid heartbeat during motor activity (catabolic phase) and a decrease in heart rate after the cessation of motor activity (phase of excessive anabolism).

In the fundamental work “Physiological mechanisms and regularities of individual development”, where the results of research performed in the laboratory of age physiology and pathology in the 1930–1980s were summarized, I. A. Arshavsky gives the following definition of the adaptive response of the organism:

Adaptation is the response of physiological-morphological transformation of the organism and its parts resulting in an increase of its structure-energy potential, that is, free energy and potential for action. This response is induced by such irritants of the environment which can be referred to as physiological, although they do demand the expenditure of an amount of energy. We call these irritants physiological because the energy expenditure resulting from them is compensated by its acquisition, that is by the functional induction of excess anabolism, automatically induced by each activity. As a consequence, a spiral-like transition of a developing organism to a higher level of potential working ability takes place. This adaptive response is characterized by three phases: the first phase is anabolic, the second one catabolic, giving the possibility of realizing another activity, and the third phase excessively anabolic. Speaking in terms of thermodynamics, we can interpret the above definition of adaptation as a response resulting in an increase of the free energy of the organism and ... the state of stable nonequilibrium... [4].

Note that the phase of excessive anabolism occurs only after activity. If there is no activity, then there is no development (and not even just recovery, since non-functioning cells self-eliminate by means of apoptosis mechanisms; an inactive organism degrades). It is in the phase of excessive anabolism that all development processes take place, in particular, the processes of long-term memory that stores our life experience. “...All motor activity, beginning with a zygote, is a factor of induction of anabolism, no matter if the motion is stimulated endogenously—due to the necessity of satisfying the demand for food, or exogenously—due to an irritant action of physiological stress. The purpose of this anabolism is not just recovery of the initial state, but necessarily excessive recovery. This transfers the system... to a new level, at which its internal energy increases... In the case of nervous cells it is manifested in trace hyperpolarization and ... in an increase in RNA content. For muscle cells it is trace hyperpolarization, trace hyperrelaxation, and an increase in

protein content. It was found that ... anabolic excess is realized only due to activity. Experimental blocking of activity causes, in spite of continuing food intake, to a temporary stop or inhibition of growth and development, or even its stasis" [4].

Though I. A. Arshavsky, speaking of growth and development, means a growing organism, including the prenatal stage of its development, all of the above concerning the obligatoriness of function for excessive anabolism refers to any living creature independent of its age. In adult individuals, growth and development continue during their entire life in the form of expansion of the volume of long-term memory. The factors controlling the development of the nervous system of a growing organism and the processes within which this development is realized, are concentrated in the brain structures in adulthood, where new distributed multilevel systems of local neuron networks are formed (neocortex columns) in response to a sequence of signals about subjectively new realities of the environment. At present, the complete unity of the biochemical processes of development of the nervous system and of memory formation has been proven.

3 Graphical Representation of Self-reproduction (Adaptation) Cycles

Following [5], let us illustrate the above graphically. When we set ourselves the task of graphically representing the normal physiological adaptive response as a cyclic process, the value of structural energy of the organism (F) is naturally assumed to be on the ordinate axis. It is suitable to quote here a comment made by I. A. Arshavsky concerning this quantity:

...the most essential qualitative feature of living systems is their state of nonequilibrium, steadily supported by the work of the structures comprising it—the work directed against their transition to the equilibrium state. The chemical energy of food substances entering the living systems can not directly transform into work. The energy, first and foremost, is used for producing free energy in the structure of the living system. In living systems, in contrast to nonliving ones, neither thermal nor chemical energy is the source of the work being performed by them, as is the case with thermo- and chemodynamical engines; the structural energy of the organism is responsible for that. The latter, in contrast to mechanical, thermal, chemical, electromagnetic, gravitational and nuclear energy is a specific form of energy inherent only in living systems [4].

Figure 1a illustrates the changes of the free structural energy during one typical cycle of self-reproduction. Point *A* corresponds to the moment when an action on the organism requiring its response took place, point *B*—the time of completion of the response, point *C*—completion of the (excessively anabolic) phase 3 of the adaptive response. The dashed line shows the minimum value of free energy of the organism when it performs the work related to its adaptive response to the irritant. The figure illustrates the case when this limit is not exceeded: the free energy of the organism at point *B* (F_B) is in excess of F_{\min} ($F_B > F_{\min}$). In phase 3, the organism returns

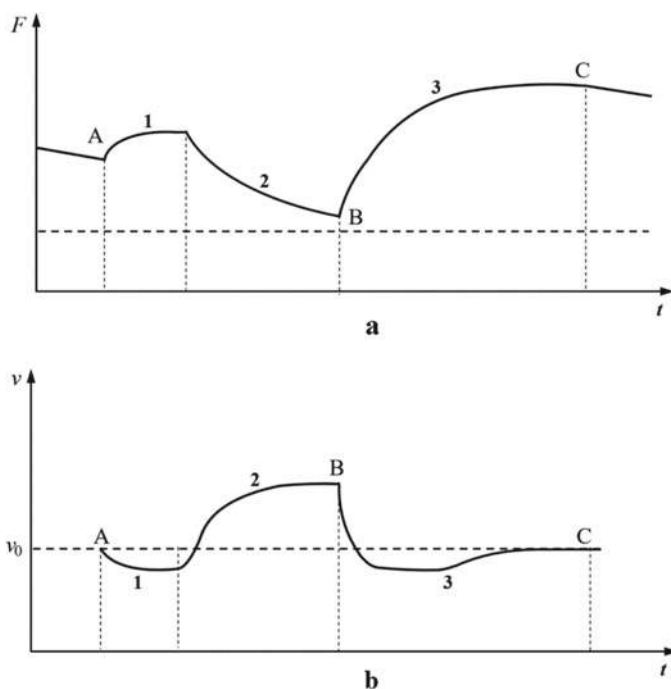


Fig. 1 **a** The change in the structural energy of an organism (F) with time (t) during an adaptive response (single cycle of self-reproduction). **b** The corresponding change in heart rate (v) with time (t)

to a state of rest (point C), with higher free energy F_C as compared with that of the initial state of rest.

Another important parameter of the cycle characterizing the deviation of the organism from the stable steady state is the heart rate (v). One argument in favor of this choice is the circumstance that a change in the performance of the cardiovascular system in the conditions of catabolism and anabolism occurs in opposite ways. In the catabolic phase, the heart rate increases. In contrast, a decrease in the heart rate (vagal bradycardia [6]) is typical of anabolism. "...Vagal bradycardia is a necessary correlate both of the first and of the third anabolic phases of the physiological response of the entire organism. For the heart, the periodically occurring vagal bradycardia plays the role of providing the growth and development of the heart muscle. This is achieved through an increase in the duration of the diastolic pause ensuring an excess of regeneration processes upon every heart diastole" [4].

Facing a new stimulus, the organism first shows a temporary decrease of the heart rate (anabolic reaction associated with the prognostic phase) which is then followed by an increase of the heart rate in the phase of action, Fig. 1b. After achieving the result, the heart rate falls enabling excessive anabolism and restoration of the

structural energy and finally comes to the equilibrium value—however, already with an increased level of structural energy.

Arshavsky's adaptive response results in the increase of structural energy of the organism:

$$\Delta F = F_C - F_A > 0$$

and, consequently, an increase in its ability of performing work under following irritants of a similar (but not only) kind. I. A. Arshavsky referred to phase 3 of a normal physiological adaptive response as a phase of “excessive anabolism”. This phase starts after performing a certain function at phases 1 and 2 by the organism and only in those structures which were involved in the realization of the function [4]. A dominating component of the adaptive response at these stages is motor activity both at the level of the whole organism (skeletal muscles), and at the cellular level [7].

4 Hyper Cycles of Self-reproduction

The cyclic nature becomes more visible if the free energy of the system is plotted as function of the heart rate, ν .

In Fig. 2, Arshavsky's adaptive response is shown schematically in the coordinate system free energy (F)—heart rate (ν). Due to excessive character of anabolic stage, this is an open cycle (“hyper cycle” according to I. A. Arshavsky's terminology).

In Fig. 2, ν_0 is the heart rate in the state of rest of a wakeful organism; point A corresponds to the time of irritant action; point B to the completion of necessary

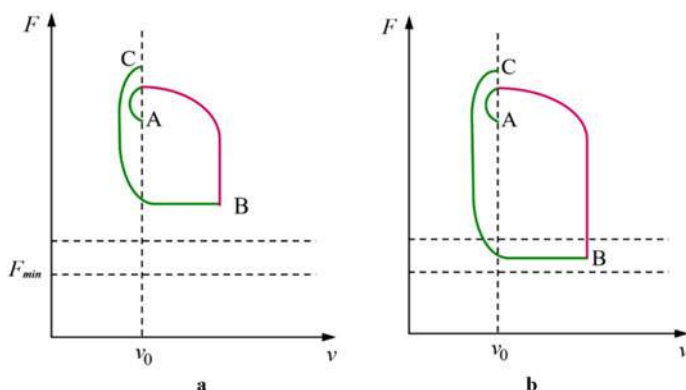


Fig. 2 Schematic of the adaptation response due to I. A. Arshavsky: F —structural energy, ν —heart rate. **a** Normal physiological response, **b** a critical physiological response

responses; point *C* to the completion of the third, anabolic, phase of the response and thereby the entire adaptive response.

Point *A* is the point at which we are caught up in the need to act. Immediately there is a decrease in heart rate and an increase in free energy in the prognostic phase, the formation of a functional system that will provide the subsequent action. Then this functional system acts (the region indicated by the red line and ending at point *B*). The action is followed by recovery, but the recovery is excessive, so that the body reaches a higher level of free energy than at the beginning. F_{\min} is the energy, below which one should not go. Why? The reason is that during the “action” phase, carbohydrates are used as energy supply. If one continues to act, ignoring the capabilities of the body, the body will have to switch from carbohydrate to lipid metabolism, and then to protein metabolism. Horizontal dashed lines mark the area where lipid-protein metabolism is starting, (this is actually a certain transition region rather than a boundary). Thus, a healthy organism can sometimes use a part of its necessary structural resources (as shown in Fig. 2b) but restore this damage completely during the following excessive anabolic stage.

Depending on the strength of external influences, this cycle can be realized in several forms. For example, if the “problem to solve” turned out to be very simple and a solution was found with a minimum expenditure of resources, then at the completion of the action, the energy will not fall below the initial one (Fig. 3a). After this, however, the recovery phase still follows. All this happens within carbohydrate metabolism. Stronger impacts have already been discussed above.

And finally, if the irritant is too strong than the body is forced to enter the sphere of lipid-protein energetics (Fig. 3b). At the end of the action, in this case, the pulse drops, but not to the equilibrium value, and after that it very slowly returns to normal. Only after this the reverse transition to carbohydrate metabolism proceeds. After this, a

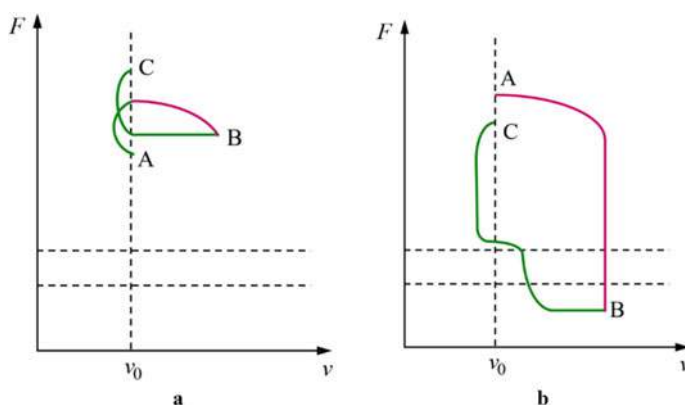


Fig. 3 Schematic of the adaptation response due to I. A. Arshavsky: F —structural energy, v —heart rate. **a** Response to weak stimuli (training reaction), **b** overcritical stimuli (pathogenic reaction—stress)

rapid decrease in pulse occurs, a transition to the stage of anabolism, which, however, may not be enough to completely restore the wasted structural energy.

5 Non-specific Adaptive Responses

The described cycles of self-reproduction are *specific* for each irritant. The particular nature of the irritant determines the image of the result of the action as well as the structures of the organism comprising the functional system formed in the phase 1 and the duration of the activity needed to achieve the result. However, not always the result can be achieved in one cycle so that several cycles may be required. In conditions of prolonged focused activity, a general *non-specific adaptive response* of the body is formed, controlled by the hypothalamus-hypophysis-adrenal gland system [4, 8, 9]. Depending on the strength of the stimulus, it may be a “training reaction” [8], an “activation reaction” [8] or an enhanced activation reaction [10]. Finally, in a situation where efforts to achieve the result are required that exceed the physiological capabilities of the body, a pathological response develops—stress [8, 11].

Apparently, one of the listed adaptation answers corresponds to the dominance of the long-term purposeful activity of each of the options for the elementary reproduction cycle described above.

The four non-specific adaptation reactions are summarized in Fig. 4, which shows the power of vital activity (consumption of structural energy per unit time) versus time.

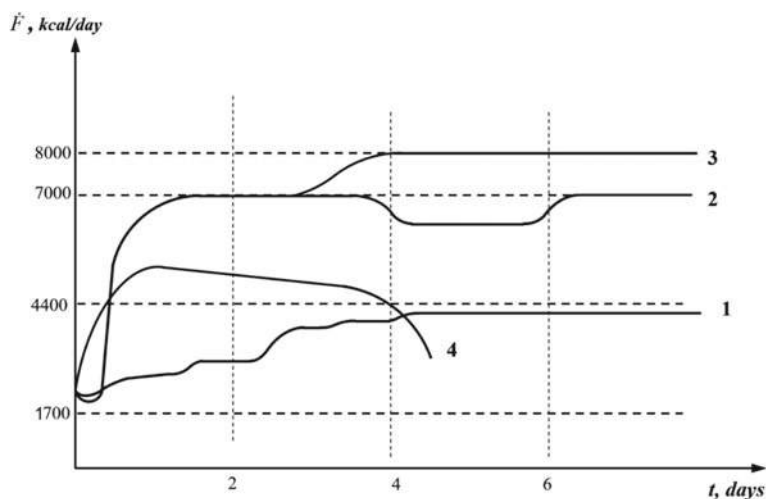


Fig. 4 Metabolic power characteristic for: 1: weak stimuli (training reaction), 2: average stimuli (activation reaction), 3: enhanced activation, 4: stress

Curve 1 corresponds to the case of weak stimuli (Fig. 3a). The maximum metabolic power developed by humans in the state of this “training response” to weak stimuli does not exceed 4400 kcal per day [12].

The second option corresponds to the so-called activation reaction [12]. Being in this state is very becoming to the organism. Changes in the immune system accompanying this response occur in the direction of increasing the organism’s resistance, improving the immune systems. Without going into detail about the mechanisms of formation of the activation response (“physiological stress” according to I. A. Arshavsky), we should note that the adaptive responses of the organism are largely controlled by hormonal activity of the hypothalamus–hypophysis–adrenal-gland system. When the organism functions in the regime of the Arshavsky’s adaptive response (activation response according to Ukolova [8]), the system releases anabolic agents into the bloodstream: growth-hormone-releasing factor of the hypothalamus, somatotrophic hormone of the hypophysis, mineralcorticoids and testosterone of the adrenal gland, etc. In general, the processes of anabolism, recovery, and development dominate in the organism in this case. This is facilitated by the fact that the activation reaction is accompanied by a certain decrease in blood coagulability, which ensures high mobility of changes in blood supply (and therefore nutrition) of all functional tissues of the organism.

During the activation response a human can use up to 7000 kcal per day [12]. For comparison, recall that the base metabolism rate required for sustaining the vital activity of the organism in a state of complete rest is approximately 1600 kcal per day, and in our ordinary vital activity, including laborious activity, we use about 3500 kcal per day.

In the case of prolonged activity, long and intense work is possible within the framework of a reaction of enhanced activation with a power reaching 8000 kcal per day.

6 Stress and Necessity Avoiding It

However, in reality, man often finds himself in situations requiring efforts that exceed the physiological capabilities of the organism. A response to extreme irritants (“strong stimuli” [8, 10]) allows the organism to survive as a living system, but leaves behind injuries and pathological changes. This modus of activity is shown in Fig. 4 as line 4.

The anabolic stage under stress is weak or is entirely absent [4] (Fig. 3b), while the catabolic stage 2 leads to expenditures of structural energy and to structural changes in the organism that exceed its regenerative abilities, the increase in structure energy at stage 3 is lower than the expenditures at the catabolic stage 2: $\Delta F = F_C - F_A < 0$. This adaptive response of animal and human organisms to excessive irritants was discovered earlier than the responses corresponding to normal physiology by Canadian physiologist H. Selye in 1936 [9]. He called the response stress (the term became widely used mostly owing to journalists who used and abused the word). The changes

in the hypothalamus–hypophysis–adrenal-gland system controlling the hormonal status of the organism, as Selye [9, 11] and other researchers have shown, are of a mobilizatory character. The hypothalamus, secreting the corticotropic-releasing factor into the blood vessels of the hypophysis triggers the release of adrenocorticotropic (ACTH) hormone by the hypophysis, which in turn induces secretion of corticosteroids by the adrenal cortex, along with the release of catecholamines, adrenaline and noradrenaline, by the adrenal medulla. Both corticosteroids and catecholamines are factors whose function is the emergency mobilization of the organism. Energy is mostly generated on the lipid (adipose) basis. Carbohydrate energy is suppressed. Physiologists even have a proverb: “Fats burn in the carbohydrate fire, but carbohydrates do not burn in the fat fire” [8]. This is a very important circumstance, since the nervous system functions only on the basis of carbohydrate energy. Therefore, while in stress, the cognitive component of vital activity is either suppressed or absent. In other words, stress actions are unacceptable in the educational space and in educational cognitive activity. When designing educational technologies, the pathological adaptive response, stress, should be considered only insofar that it must be eliminated.

Let us note some other features of the pathological adaptive response. An increase in coagulability occurs (a natural evolutionary adaptation: stress situations may result in bloodshed). Blood circulation becomes less dynamic, tissue nutrition is decreased or stopped, including that of functional tissues. One of the characteristic stress symptoms is the formation of multiple bleeding ulcers in the digestive system [8]. Thus, stress never does without bleeding.

The immune system is literally switched off. Its major organ, the thymus, drastically decreases in size, all immune factors are released into the bloodstream, and as a result the organism is relatively resistant to pathogenic influences for three days. But in the stage of exhaustion which sets in on the third or fourth day of stress, the human becomes ill. And although the “weak links” of the organism differ from person to person, in most cases acute tonsillitis occurs, due to the defenselessness of the tonsils, which are lymph-glands, and therefore, a part of the immune system. In any case, if one finds that the mandibular glands are swollen, one should try to remember what trouble happened 3 days ago, and by revealing the stressor, eliminate it, changing the attitude towards it or by restructuring near-term plans.

It should be noted that under stress the water-mineral metabolism changes so much that the weight of a human increases by 2 kg in several hours [12]. Therefore, a person who regularly observes their weight can easily discover stress by weighing.

7 Conclusions

We provided a brief review of adaptation reactions stressing the importance of their cyclic nature as well as the first anabolic phase of a cycle (prognostic phase, when the *image of the result of action and the action itself* is created). The central role in the self-reproducing cycles is played by the anabolic phases. Normally, one of the

anabolic phases follows a catabolic phase during which the resources of the organism are expended. However, in exceptional cases, it is possible that the whole activity remains anabolic.

We discuss the importance of avoiding stress reaction—both in terms of health and educational activities (as stress never leads to development of long-term memory).

In the past years, the author discussed some of this topics with S. G. Psakhie in context of creative scientific activity and its organization.

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Seeing What Lies in Front of Your Eyes: Understanding and Insight in Teaching and Research



Elena Popova, Valentin L. Popov, and Alexander E. Filippov

Abstract In the present paper, we considered the phenomena of understanding and discoveries (as a sort of “social understanding”) and found that the empirical properties of these phenomena (the critical character and emerging of a new property) have much in common with first-order phase transitions. From this point of view, we discuss both the process of understanding and discoveries and the reasons impeding “seeing what lies in front of our eyes”. In our opinion, these ideas can be further studied on the same phenomenological basis, without detailed understanding of the underlying neuronal mechanisms.

Keywords Understanding • Insight • Discovery • Phase transitions • Order parameter • History of science • Friction • Contact • Adhesion

1 Introduction

This paper is devoted to a phenomenon of fundamental importance for many areas of human activity, including research and teaching—the phenomenon of *understanding*. It is clear that teachers must ensure that the material they mediate is *understood* and not only memorized or “only” practiced well. However, this central phenomenon of learning always remains a mystery. The act of understanding can be compared to that of “seeing”: When understanding something, we suddenly see “the picture as a whole”. However, this is not at all easy—as Goethe once said: “The hardest thing to see is what lies in front of your eyes”.

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Based on historical examples, we suggest considering the mechanism of “sudden understanding” as a first-order-phase transition in the cognitive space (individual understanding) or a combination of cognitive and real space (discoveries).

2 Phenomenon of Understanding

The paradoxical truth expressed by Goethe is valid both for the scientific perception of the world and for everyday life. Consider the painting “Gartenlokal an der Havel—Nikolskoe” (Fig. 1).

When people stand in front of this painting, showing a Cafe at one of the lakes in Berlin, they usually do not see anything strange in it—even if they study the painting attentively. There are just people sitting in a cafe and speaking to each other. What goes unnoticed is the fact that none of the chairs in this painting has legs: Everyone is sitting in the air! It is still a matter of debate whether Liebermann just did not have enough time and the painting is incomplete or this was his joke with the observer, but the fact is that the majority of spectators are not at all disturbed by the missing details. They just do not see that they are missing!

In this case, the details we do not see are not important for the correct perception of the painting. However, sometimes it is exactly the details we overlook that are important. In the same way, something that may be hard to understand initially, can



Fig. 1 Max Liebermann, Garden Restaurant on the Havel—Nikolskoe, 1916, Berlin, Nationalgalerie

Problem 1. Two tractors pull a box with ropes (see picture). The velocities of the tractors are directed along the ropes and are \mathbf{v}_1 and \mathbf{v}_2 . The angle between the ropes is α . What is the velocity of the box and how is it directed?

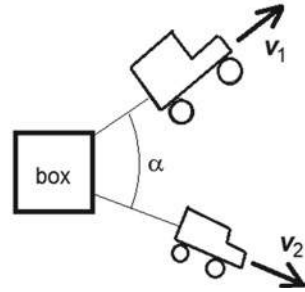


Fig. 2 A task from the “Colloquium Advanced Mechanics” at the TU Berlin

become trivial once seen from the right perspective: The solution lies in front of our eyes, but we do not *see* it.

One of the authors of this paper (V.P.) offers a course at the TU Berlin that is specifically designed to illustrate and to train the ability to understand: the “Colloquium for Advanced Mechanics”. The typical feature of the tasks that are handled in the Colloquium is that they are “very difficult problems, that are very easy to solve”. The statement is contradictory only at first sight. The tasks of the Colloquium often cannot be solved by students, but only because they do not approach them from the correct perspective.

Here is an example of a task from the Colloquium (Fig. 2). Two tractors pull a box with ropes. The velocities of the tractors are directed along the ropes and are v_1 and v_2 . The angle between the ropes is α . What is the velocity of the box and how is it directed? The task sounds simple. It is clear to everybody that it must have a solution: If you pull a box, it will have to move somewhere. And yet it is a task that even scientific collaborators at the University cannot solve immediately. The first idea that comes to mind is to sum up the velocities of the tractors. However, at least after considering the special case of parallel ropes and equal velocities, one can see that this idea is unfortunately completely wrong.

Experience shows that only exceptional personalities can solve this problem quickly and correctly—in spite of the fact that the task is solvable in two lines! There is just nothing to “calculate” in this problem! The only thing one needs is to look at the problem from the correct point of view. Once one understands the underlying principle, one can only say: “Ah!” And that *is* the solution.

The colloquium essentially consists of tasks of this kind, in which one finds the solution not by “working long and hard”, but by real *understanding*. As Goethe has said, the hardest thing to see is what lies in front of your eyes!

3 Discoveries in the History of Science as “Seeing the Obvious”

A similar phenomenon of understanding is known in the history of science. History is full of examples where great discoveries were nothing more than seeing what lies in front of your eyes [1].

Let us take fracture theory as an example. In 1921, Alan Griffith put forward an idea why the strength of materials is much lower than the theoretical one and why it depends on the size of engineering parts [2]. His paper was the beginning of the theory of fracture. But what did Griffith really do? He only said that the crack tip is in equilibrium if the change of energy due to a small displacement is zero. This energy change consists of the relaxation of elastic energy and the work of separation of surfaces. One of them is positive and the other negative. If their absolute values are equal then the crack is in equilibrium. This is of course nothing else than the principle of virtual work for mechanical systems. This principle is by no means an invention of Griffith, it was already known to d'Alembert and Leonard Euler. It is a standard, well-known equilibrium condition, which can be found in any introductory textbook on mechanics. New was only that Griffith applied this old principle to the crack. Interestingly, he did not even have to carry out any complicated analytical calculations, because the elastic energy, which is released due to a small movement of a crack tip, was already known. The corresponding problem was solved in 1911 by Inglis [3], and the expression for the work of adhesion is trivial: it is the product of the specific work of adhesion and the new surface area produced by the crack opening. Thus, Griffith equated two contributions, which were both known at that time and thus produced the famous, classical results of fracture theory!

50 years later, Johnson, Kendall and Roberts (JKR) published their famous paper on adhesion, one of the most cited papers in the field of contact mechanics [4]. In their paper, JKR note that their approach is equivalent to that of Griffith. They write: “...the approach followed in this analysis, is similar to that used by Griffith in his criterion for the propagation of a brittle crack.” As a matter of fact, JKR realized that the adhesive contact is the inverted Griffith crack (in the crack the discontinuity is mostly inside and in the “adhesive contact” outside). After that, JKR applied the principle of energy balance exactly in the way Griffith did. The only difference is in the expressions for the elastic energy. Griffith used the energy of an “internal crack” provided by Inglis and JKR used the expressions provided by Hertz [5] and Boussinesq [6]. The realization of the equivalence of the problem of adhesive contact to the problem of the Griffith crack was the main contribution of JKR, as a matter of fact, their *only* contribution. Already Griffith had all the necessary ingredients for the solution of the adhesive contact problem, he merely had no need to solve it. If he had, we probably would have had the “JKR”-solution already in 1921. It is surprising that 50 years were needed just to recognize the equivalence of these two problems!

But the history of adhesion does not end here. One of the most effective numerical simulation methods in contact mechanics is at the present the Boundary Element Method (BEM), which allows simulation of arbitrary contact configurations [7].

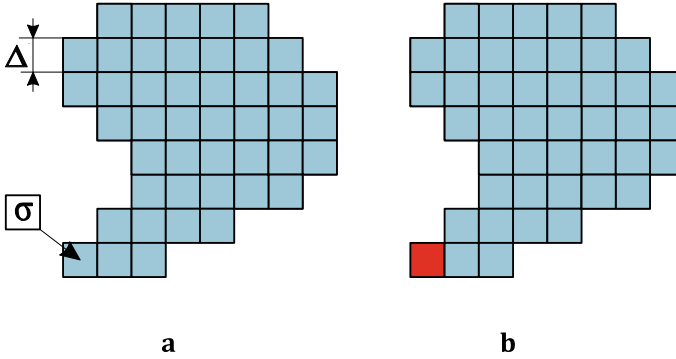


Fig. 3 In each calculation step, stress in each particular discretization element is determined. If the stress σ in a given element at the boundary of the contact area exceeds the critical value (3), it is “detached” and the stress in this element is set zero

However, only in 2015 an idea was put forward on how to simulate adhesive contacts in the framework of BEM: The use of a simple, unmodified Griffith criterion for the BEM detachment condition [8]. Let us explain this in more detail.

In each iteration of the BEM simulation, the stress in each discretized cell is determined and then it is decided whether this cell is still in contact or not. For non-adhesive systems, the answer is very simple: The pressure must be positive. But if you have an adhesive contact then the pressure can become negative and we have to find a criterion for when a particular element will lose the connection to the counter-body. In 2015, two scientists suggested applying the Griffith criterion: Let us assume that the contact of a cell is lost, then the stress in this cell vanishes (Fig. 3). It is easy to calculate analytically what amount of elastic energy is relaxed due to the stress vanishing:

$$\Delta U_{el}(\tau) = \kappa \frac{\sigma^2}{E^*} \Delta^3 \quad (1)$$

with

$$\kappa = \frac{2}{3\pi} \left(1 - \sqrt{2} + \frac{3}{2} \log \left(\frac{\sqrt{2} + 1}{\sqrt{2} - 1} \right) \right) \approx 0.473201. \quad (2)$$

The element is in the state of indifferent equilibrium if the change of elastic energy is equal to the work of adhesion needed for creating the free surface with the area Δ^2 , $\Delta U_{adh} = \gamma_{12} \Delta^2$, or with (1), $\kappa \sigma^2 \Delta^3 / E^* = \gamma_{12} \Delta^2$, where γ_{12} is the work of adhesion per unit area. For the critical detachment stress they obtained

$$\sigma_c = \sqrt{\frac{E^* \gamma_{12}}{0.473201 \cdot \Delta}}. \quad (3)$$

It is now assumed that if this elastic energy is equal to the work of separation then the element will detach. In other words, as soon as the elastic energy is enough for creating new surfaces, they will be created. This solution is so simple and straightforward that one can only ask why it took another 44 years to apply the Griffith criterion in numerical simulation!

The present paper is not devoted to any detailed substantiation of ideas, let us just mention that the simple criterion presented above reproduces all known analytical solutions exactly [9].

4 Understanding as Changing the Point of View

The above examples illustrate very clearly the idea that the real mechanism behind new theories and paradigms is often a transfer of knowledge from one subject area to another [1]. A similar process in individual perception is *changing the point of view*. But what exactly does that mean? This mechanism reflects fundamental philosophical roots of the term “meaning” that were elaborated in detail in the second half of the twentieth century, owing to the development of system analysis and increased interest in the nature of complicated systems. According to system analysis, properties of a system cannot be understood by looking on it from inside the system [10]. Recall the definition of a system given by R. Ackoff: “A system is a whole which is determined by its function in the system, a part of which it is” [11]. Therefore, understanding of the meaning, purpose, and objectives of any system can be achieved only at the level of the containing system (supersystem). This means that true understanding implies the placing of a given object into a corresponding supersystem where it plays a specific role. Depending on the supersystem, the same object can have different meanings.

The above historical examples have one thing in common: The “seeing” occurs by exchanging the supersystems. Very often, one hears that scientists and engineers have to “think outside the box”. From the point of view of system analysis, thinking outside the box becomes a generic, compulsory prerequisite to any true understanding.

Of interest is R. Ackoff’s comment on the relationship between the analysis of a part of a system and the supersystem: “Note that an analysis starts with division of a subject into parts and gives knowledge. Synthesis starts with combination of things and gives understanding. Analysis is the way the scientists investigate. Synthetic thinking is manifested in designing... There is one more extremely important aspect of synthetic thinking. The systems we deal with are becoming increasingly sophisticated. Scientists seek the effective ways of treating them. Unfortunately, most of them approach the problem analytically. As a result, they introduce so many variables and relations between them that we cannot cope with them. However, if complexity is dealt with in a synthetic way, by designing, like designing a high-rise building or a city, there is no level of complexity that we cannot effectively cope with” [10, 11].

If we accept the viewpoint of Ackoff, then we have to accept that understanding is the final result of a synthetic activity aimed at a broadening the context in which the

given object is considered or at trying a large variety of “contexts” until a “correct” one is found.

There is another potentially important point of this process, which is best illustrated not on examples of personal perception but on the history of discoveries: It is not always clear that the correct supersystem has already been found, so that the final solution may require many attempts. In the history of science, this is seen by the very widespread phenomenon of multiple discoveries.

5 Multiple Discoveries

Very often scientific results are obtained multiple times—“re-discovered” without knowing the predecessors. Take as an example the history of the so-called Method of Dimensionality Reduction [12]. This is a method of representing contact mechanics in a simple way that can be taught even to undergraduate students and can be used by any practical engineer. It was developed in the decade from 2005 to 2015. However, it was invented already much earlier and reinvented many times in the course of history. And each time, people just did not see what was lying in front of their eyes. It was first formulated in a paper by Schubert in 1942 in Germany [13]. The same solution was found by Galin in 1946 in Russia [14] and later by Green and Zerna in 1954 [15]. Sneddon translated the book of Galin from Russian into English in 1950th (see a later publication [16]) and also published 1965 a very influential paper which contains exactly the same solution, since then mostly known as the Sneddon solution [17]. In 1998 Jäger suggested an alternative physical interpretation of the same equations [18]. Finally, they were reformulated as a simple mnemonic rule and generalized to a large variety of contact problems in 2005 to 2015, thus creating MDR [10, 12]. It is striking that it took about 70 years to realize that the solution lies in front of our eyes!

6 Understanding as a Phase Transition

The sudden character of understanding shows that this is a critical phenomenon. A prerequisite for it is a synthetic (design) work, but the final act is often perceived as “instantaneous”. Understanding is bringing order in a large number of elements. It can thus be interpreted as a “phase transition” establishing a “long range order” in the corresponding “cognitive space”. This transition can be easily followed phenomenologically by considering an impressionist painting from various distances. When considered in the vicinity, the painting looks like a chaotic set of color spots (Fig. 4). Going further from the picture, one suddenly recognizes the picture as a whole. Interestingly, this process is almost reversible so that one can repeat this “act of understanding” or “act of recognition” many times. At this point, we cannot discuss the question of the exact neuronal mechanism of this phase transition. We just use



Fig. 4 Painting “Lisette Sewing in front of the Entrance Door of Marquerol” by Henri Martine: **a** view from vicinity, **b** view of the whole picture. Tel Aviv museum of Art

the finding of the theory of phase transitions, which says that local interactions in a distributed system can lead to a “sudden” establishment of a long-range order in the system when interactions achieve a critical value [19]. This change can lead to the appearance of qualitatively new properties, such as superconductivity. In the human perception we see empirically similar sudden state changes with the emerging of a new property—that of “understanding”. It surely would be an intriguing and fruitful task to try to find out what exactly the neuronal mechanisms of this transition are, which could greatly facilitate the development of didactics.

The analogy of understanding and seeing brings another important aspect into play.

7 Interrelation of Personal Understanding and Discoveries

The fastest runner doesn't always win the race,
 and the strongest warrior doesn't always win the battle.
 The wise sometimes go hungry,
 and the skillful are not necessarily wealthy.
 And those who are educated don't always lead successful lives.
 It is all decided by chance,
 by being in the right place at the right time.
 Ecclesiastes 9:11

While the acts of personal understanding and scientific discovery have much in common, one has to also see the essential difference between both. The above example of the history of the MDR shows that “understanding” in a historical context does not merely mean that something is seen by an individual scientist. It is important

that the act of seeing a scientific result coincides with the act of public recognition of its importance: Understanding “not at the right time” “does not count”. Only “collective seeing” is a true discovery. We often do not really know how many times the same scientific truth was re-discovered and intensive historical research is needed to answer the question: “who was the first”? Maybe one can speak of the act of “public understanding”, which means understanding reaching a large (relevant) community. However, it is very difficult to know in advance whether a particular result will be broadly accepted by scientific community. That is why it is so difficult to make “discoveries on demand”—in reality, discoveries are in most cases understood as such a posteriori—in rare lucky cases of (later) public recognition.

The analogy with phase transitions leads to the conclusion that “authorship” of particular discoveries is relative. If the medium is far from the critical state, no fluctuations will create the long order parameters, and vice versa, if it is in the critical or overcritical state, then *any* nucleus will lead to the phase transition. Can we blame those researchers who by chance did not work in some field which the future would show to be very important and, on the contrary, praise those who for some reason were “centuries ahead of their time” (in most cases without knowing this)?

For success of a personal scientific career, it is very important to work on problems which may have a public resonance (at least in the relevant scientific community). The highest scientific qualification lies exactly in the general orientation of what problems are “of interest”.

However, for the progress of science as a whole, this personal success is of no relevance. The true precondition of phase transitions is the local interaction. Preparation of this local interaction is as important for science as being an “ingenious inventor” (the latter often meaning to be a random “fluctuation” at the right time and at the right place.)

8 What Prevents Us Seeing What Lies in Front of Our Eyes?

The analogy of the phenomenon of understanding with phase transitions suggests a solution to the question of what prevents us from seeing what lies in front of our eyes. The fact that the “act of understanding” often occurs suddenly, implies that the underlying phase transition is a first-order phase transition. This means that the transition always occurs through formation of initially small nuclei, which later expand to the whole phase space. This propagation is connected with some “friction” in the boundary line. A very similar phenomenon takes place in social processes. A society of sufficiently large volume (including the scientific society as well) causes a practically infinite barrier for the immediate acceptance of new knowledge, even if it was already achieved by an individual researcher. Thus, the factor impeding the

understanding is a resistance to the change of state. This resistance is a generic property of any first-order-phase transformation. What is sometimes called the “inertia of thinking” is in reality a “friction of thinking”. The essential question is of course what factors determine this “friction”. Just as for example ferromagnetic materials can have a large or small internal friction for the motion of phase boundaries (corresponding to hard and soft magnetic materials), the resistance to understanding may be higher or lower depending on factors which are not yet understood. Their determination could greatly facilitate didactics.

Similarly, discoveries can be considered as acts of “social understanding”. They also occur “suddenly” (on the time scale of historical processes) and thus have the features of first-order transitions. The phase space now contains both “cognitive dimensions” and the real space dimension (in form of a geographic distribution of researchers). Any nucleus of new understanding will have to overcome the frictional force for boundary propagation. In everyday language, one can say that anybody offering a radically new idea will stumble upon a consolidated resistance by the whole system of knowledge accumulated and lovingly ordered on the shelves.

In the theory of first-order transitions it is known that a high boundary friction leads to the necessity of “overheating” the system to initiate the transition. In our analogy with understanding, this means that one has to do much more work and to accumulate more information related to the particular topic in the space of knowledge than it is necessary to perform “act of understanding” itself.

Another property of the first-order transition is the existence of the threshold of “absolute instability”. In our analogy, this means that at some point it becomes impossible to ignore new knowledge and the transition becomes unavoidable.

The nuclei of a new phase can also dissolve again if the systems moves away from the critical point. The history of civilization provides many examples of the reverse process when already achieved knowledge “suddenly disappears”, if for some reason it loses its social importance.

9 Conclusion

In the present paper, we considered the phenomena of understanding and discoveries (as a sort of “social understanding”) and found that the empirical properties of these phenomena (the critical character and emerging of a new property) have much in common with first-order phase transitions. In our opinion, these ideas can be further studied on the same phenomenological basis, without detailed understanding of the underlying neuronal mechanisms—similarly to the famous treatment of the theory of superconductivity in the phenomenological theory of phase transformations [20].

However, just as a statistical physics give a deeper insight into the physical understanding of phase transitions, understanding the neuro-physiological mechanisms of perception and handling of “newness” by our brain would strongly facilitate also the phenomenological view on understanding. Some important findings in this relation can be found in [10].

In philosophy, this picture has already been developed over centuries under the notion of “measure” as a unity of quantity and quality (Hegel) [21]. However, we find that the picture provided by the theory of phase transitions provides more details and more understanding of the underlying processes.

The true understanding is the only quality, which enables a person or a community to make real progress in any branch of science or engineering. The analogy of understanding to a long range order provides an (we hope useful) illustration of what sometimes is called “complete knowledge” [10], meaning that understanding implies seeing the “picture as a whole”. The long range order is exactly what provides a picture the “wholeness”.

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