

Simulation of rock massif tension at ore underground mining

Vladimir Golik

*ScD, Professor,
North Caucasus mining metallurgical Institute (State Technical University)
Russia*

Vitaly Komashchenko

*ScD, Professor,
The Peoples' Friendship University of Russia*

Vladimir Morkun

*Vice-Rector for research, ScD, Professor,
Professor of Computer Science, Automation and Control Systems Department,
Kryvyi Rih National University,
Kryvyi Rih, Ukraine*

Olga Burdzieva

*Candidate of Geographical Sciences,
Geophysical Institute of Vladikavkaz Scientific Center of RAS,
Russia*

Abstract

In the article the stresses in the elements of geomechanical system were determined. Methodology of research organization, massif management options and massif state control parameters optimization are described. Rock massif tension at ore underground mining is simulated.

Keywords: ROCK MASSIF, TENSION, ORE, UNDERGROUND MINING

In roach deposits developing a combination of induced geomechanical processes with natural processes violates the geodynamic equilibrium in upper crustal and activates the catastrophic events [1-5]. Stability of ore-containing massifs is determined by the level of stresses on the contour of stope ores, which is governed by voids filling with hardening mixtures after the evidence of this option effectiveness, for example, the method of photoelasticity [6-8]. The most complex is the mining of heavy ore deposits by combined opencast and underground methods. The criterion of combination effectiveness is the preven-

$$\sigma_1 - \sigma_2 \geq \sin \delta (\sigma_1 + \sigma_2) + \sigma_{rs} + (1 - \sin \delta), \quad (1)$$

where σ_1, σ_2 - are the stresses in contour point; δ - is the angle of internal friction, 30° ; σ_{rs} - is the rock strength.

In-situ stress:

$$G_H = \gamma H \frac{G_M}{\sigma_{in}} \quad (2)$$

where γ - is the ore and host rocks density, t/m^3 ; H - is the stratification depth of the point from the surface, m ; σ_{in} - is the stress in the model; G_M - is the stress in the model, MPa ; G_H - is the in-situ stress, MPa .

For determining of stresses in the model the following expression is used:

$$\sigma_M = \sigma^{1.0} \cdot n, \quad (3)$$

where $\sigma^{1.0} = 0,1 \text{ kgf/cm}^2$ per one band; n - is the band number in point of interest of model.

Stresses in the model and in-situ are determined from the expression

$$G_H = \gamma H \frac{G_M}{k}$$

where k - is the similarity coefficient.

Condition of massif was investigated under conditions:

- horizontal stress 0,5; 1,0; 1,5;
- the force vector inclination angle to the vertical axis $\alpha = 0$ for each value of horizontal stress;
- large fill modulus $E = 0,1 \text{ MPa}$, host rock modulus - $1,4 \text{ MPa}$;

tion of critical stresses [9-13]. Technique for organization of research includes the selection of the optically active materials; development of a device for patterns loading at different angles of force vector inclination based on the lateral thrust; results photoregistration devices [14-17]. Models were made from the optically active polyurethane with fringe value of $7,6 \text{ MPa}$ for conditions: laying depth of mine working from the surface is $350m$, the volumetric weight of overlying rocks is $3,0 \text{ t/m}^3$. The stability of a given contour point is described by the condition

- options with cameras large fill and without it.

The options of massif control are characterized by stresses values, which are measured in cameras, interchamber pillars and on the vertical section of the camera.

For a coefficient of horizontal stress $\lambda = 0,5$ the maximum stresses in arch keystone zones and camera walls are equal to $7,6 \times 7,5 = 57 \text{ MPa}$, and in arch pillar apex to $7,6 \times 2 = 15 \text{ MPa}$. The maximum compression stresses in interchamber pillar are $7,6 \times 6,5 = 49 \text{ MPa}$.

For a coefficient of horizontal stress $\lambda = 1,0$ the stresses in arch keystone zones, camera walls and in arch pillar apex are equal to $7,6 \times 6,5 = 49 \text{ MPa}$. In pillar the maximum stresses are reduced to $7,6 \times 5,5 = 42 \text{ MPa}$.

For a coefficient of horizontal stress $\lambda = 1,5$ the stresses in arch keystone zones and camera walls are equal to $7,6 \times 6,5 = 49 \text{ MPa}$, and in arch pillar apex to $7,6 \times 8,5 = 64 \text{ MPa}$ in contrast to 15 for coefficient of horizontal stress $\lambda = 0,5$.

The stresses in arch pillar was:

- for a coefficient of horizontal stress $\lambda = 0,5$ $7,6 \times 5,5 = 41 \text{ MPa}$;
- for a coefficient of horizontal stress $\lambda = 1,0$ $7,6 \times 13,5 = 102 \text{ MPa}$;
- for a coefficient of horizontal stress $\lambda = 1,5$ $7,6 \times 18,5 = 140 \text{ MPa}$.

The maximum stress at the camera contours and keystones of arch pillar are developed with a coefficient of horizontal stress of 1,5 (Table 1).

Table 1. The stresses in the elements of geomechanical system, MPa

Thrust coefficient	Open mined-out area	Filled with hardening mixture
Arch pillar of block		
0,5	3	2
1,0	7	5
1,5	13	9
Left arch keystone		

0,5	5	6
1,0	4	5
1,5	3	4
Right arch keystone		
0,5	5	5
1,0	5,5	6,5
1,5	6	8

Optimization of the massif state control parameters is often a decisive factor in ensuring the efficiency of deposits development [1, 5, 7, 18].

Conclusions

The level of technogenic stresses is determined by simulation on low molecular materials with results photodetection. The most stress have an arch pillar of cameras. Large fill of cameras reduces the stress level up to 2 times. In options without large fills in interchamber pillars the stress concentration is close to critical.

References

- Amvrosov A.F. (2014). *Monitoring opasnykh geologicheskikh protsessov pri nedropolzovani* [Monitoring of dangerous geological processes in the subsoil use], *GIAB*, No7, pp.45-50.
- Logachev A.V. (2013). K voprosu o geotekhnologicheskikh variantakh poetapnoy razrabotki mestorozhdeniy [On the issue of geotechnical variants by phased development of deposits]. *Tsvetnaya metallurgiya [Non-ferrous metallurgy]* No4, p.p.46-50.
- Zuev B.Yu. (2014). Fizicheskoye modelirovaniye geomekhanicheskikh protsessov v blochno-iyerarkhicheskikh massivakh na osnove yedinogo kompleksnogo usloviya podobiya [Physical modeling of geomechanical processes in block-hierarchical massifs based on a single integrated similarity condition], *GIAB*, No4, p.p. 67-73.
- Shestakov V.A., Shalyapin V.N., Litovchenko T.V. (2005). Teoriya optimizatsii i sovershenstvovaniya podzemnoy razrabotki slozhnykh rudnykh zalezhey [The theory of optimization and improvement of complex ore deposits underground mining]. Novochoerkassk: SRSTU (NPI).
- Golik V., Komashchenko V., Morkun V. (2015). Feasibility of using the mill tailings for preparation of self-hardening mixtures. *Metallurgical and Mining Industry*, No3, pp. 38-41.
- Lyashenko V.I. (2001). Sovershenstvovaniye dobychi poleznykh iskopayemykh kombinirovannymi sposobami vyshchelachivaniya [Improving of mineral extraction by combined leaching processes], *Mining Journal*, No1, p.p. 9-14.
- Golik V., Komashchenko V., Morkun V. (2015). Innovative technologies of metal extraction from the ore processing mill tailings and their integrated use. *Metallurgical and Mining Industry*, No3, p.p. 49-52.
- Mindeli E.O., Kusov N.F. Korneyev A.A., Martsinkevich G.I. (1978). Kompleksnoye issledovaniye deystviya vzryva v gornykh porodakh [A comprehensive investigation of the effects of the explosion in the rocks]. Moscow: Nedra.
- Sekisov G.V., Rasskazov I.Y. (2014). Creation of a research and production mining and processing complexes for innovative supporting of mining industry. *GIAB*, No 9, p.p. 113-121.
- Morkun V., Morkun N., Pikilnyak A. (2015). Adaptive control system of ore beneficiation process based on Kaczmarz projection algorithm, *Metallurgical and Mining Industry*, No2, pp.35-38.
- Kachurin, Vorobev S., Shkuratkiy D., Bogdanov S. (2015). Environmental Danger of Worked and Liquidated Coal Mines Open Areas. 5th International Symposium. Mining and Environmental Protection, Vrdnik, Serbia, pp. 141-149.
- Kachurin N. M., Efimov V. I., Vorobev S. A., Shkuratkiy D. N. (2014). Evaluating of closed mines mining lease territories environmental safety by gas factor, *Eurasian Mining*, No2. pp. 41-44.
- Rakishev B.R. (2013). Complex usage of ore in the enterprises of non-ferrous metallurgy of Kazakhstan. *Gorniy Zhurnal*, No 7, pp. 67-69.
- Kantemirov V.D. (2014). Technologic features of the development of new raw material bases. *GIAB*, No6, p.p. 369 – 373.
- Kornilkov S.V., Jakovlev V.L. (2015). O metodologicheskoy podhode k issledovaniyam v oblasti osvoeniya nedr na osnove sistemnosti, kompleksnosti, mezhdisciplinarnosti i innovacionnoy napravlenosti. [On the methodological approach to research in development of

- mineral resources on the basis of a systematic, integrated, interdisciplinary and innovative orientation], *Gornyj zhurnal*, No 1.
16. Brotanek I., Voda I. (1983). *Konturnoye vzryvaniye v gornom dele i stroitelstve* [Contour blasting in mining and construction]. Moscow: Nedra.
17. Drukovanny M.F., Kuts V.S. Ilin V.N. (1980). *Upravleniye deystviyem vzryva skvazhinnykh zaryadov na karyerakh* [Control of borehole charges blasting action in open pits]. Moscow: Nedra.



Complex approach to implementation of filling emulsion explosives Ukrainit in underground conditions

Ihor Kovalenko

*Ph.D., Assoc. Professor
Ukrainian State University of Chemical Technology,
Dnipro, Ukraine
E-mail: il-kovalenko@mail.com*

Nikolay Stupnik

*D.Sc., Prof.,
Kryvyi Rih National University,
Kryvyi Rih, Ukraine
E-mail: knu@alba.dp.ua*