Review paper http://dx.doi.org/10.59456/afts.2023.1528.001M

CHOICE OF EXCAVATION METHOD OF THE ORE DEPOSITS

Majstorović Slobodan¹, Tošić Dražana¹, Torbica Duško¹

¹University of Banja Luka, Faculty of Mining Prijedor, <u>slobodan.majstorović@rf.unibl.com</u>

ABSTRACT

The paper defines choice of the optimal excavation method for ore deposits, which are characterized by general irregularity due to their origin, occurrence and different content of the usable minerals. In such complex conditions, the choice of the excavation method is defined according to: the natural characteristics of the deposit and according to the techno - economic parameters of the comparison methods and the methods of multi-criteria optimization.

Keywords: excavation method, natural characteristics, multicriteria optimization, AHP method

INTRODUCTION

The choice of the optimal manner of excavating ore deposits is the most important phase when designing the future of underground mine. From the selected manner of excavation (methods of excavation) depend on [1]:

- the economic indicators of mine operation, (the cost of deposit exploitation is 60% of the total costs of the underground mine)
- work safety of employees and equipment
- use of certain mining mechanization

The choice of the excavation method of ore deposits is a complex and very responsible procedure when designing the underground exploitation of a deposit, and the selection process itself is done in two basic manners:

- the choice of the excavation method according to the natural characteristics of the deposit is a rational manner of choosing the excavation method
- optimal manner of choosing the method of excavation, which is done on the basis of technical and economic parameters by economic comparison of variants of excavation methods, i.e. by the method of multi-criteria optimization

CHOICE OF EXCAVATION METHOD ACCORDING TO THE NATURAL CHARACTERISTICS OF DEPOSIT

The choice of the excavation method for a certain non-layered (ore) mineral deposit is made according to certain mining and geological characteristics of that deposit, whereby these characteristics can be divided into two groups:

- 1. The basic or constant factors:
 - stability of the ore and the surrounding rock
 - capacity (thickness) of the ore body (deposits)
 - dip angle
- 2. The variable factors:
 - deposit size per strike and dip
 - ore body morphology
 - value of ore
 - the character of the useful minerals distribution in the ore body
 - tendency of ore to compaction (hardening), oxidation or self-ignition
 - the hydrogeological characteristics
 - the need to preserve the surface of the terrain above deposit

The basic or constant factors are taken into account in each case, while variable factors are taken as limitations in certain cases [2].

The stability of ores and surrounding rocks defines the possibility of applying excavation methods with gape excavated areas, methods with support or methods with caving [3]. Depending on the stability of the ore and surrounding rocks, the excavation method is chosen (Table 1), as well as the parameters of the roof control system that provide safe working conditions [4,5,6].

Stability combination of cases ores	The possible groups or subgroups of excavation methods.
and surrounding rocks	
Stable ore and surrounding rocks	All groups of excavation methods are acceptable, except for excavation
	methods with caving.
	From the group of excavation methods with backfilling, a subgroup of
Stable ore and unstable surrounding	methods with the horizontal levels is acceptable, it is also possible to use
rocks	the subgroups of excavation methods in sloped levels.
	Acceptable group of excavation methods with caving, except for block
	caving methods.
Unstable ore and stable surrounding	Subgroups of excavation methods with support of excavation, then with
rocks	support and backfilling of stopes are acceptable.
Unstable ore and souronding rocks	Subgroups of excavation methods with stope support are acceptable.

Table 1. Mining methods according to the stability of the ore and the surrounding rock

Table 2 gives the choice of the excavation method depending on the dip angle and the thickness (capacity) of the deposit.

T-1-1- 7	E			4 - 41			1
I anie 7	Excavation	methods	according	to the	canacity	and din	andle
1 uoic 2.	Lacuvation	mounous	according	to the	cupacity	und unp	ungio
			0				0

Possibility and dip angle of deposits	Methods with natural maintenance of open stope (open stope methods)	Methods with ore and surrounding rock caving	Methods with artificial support of stope (methods with support)
The deposits are steep	The chamber excavation system - at any thinkness. Shrinkage stoping where the thickness must not be less than 1 - 1.3 m to avoid ore bind.	Block caving methods for thick deposits. Sublevel caving methods at deposit thickness less than 3 m.	Excavation methods with backfilling in horizontal and inclined levels - at any deposit thickness. Self-filling methods of narrow deposit. Methods with support and backfilling at every deposit thickness.
The deposits with slightly slope	Frontal and room- pillar excavation methods at medium and low thickness. Chamber methods with ore shrinkage in more thick deposits.	Block caving methods in thick deposits. Methods with caving of thick deposits.	Methods of excavation horizontally and inclined with backfilling. Support methods. Methods with support and backfilling of stopes.

Table 3. presents systematized basic or constant factors and their influence for certain groups and subgroups of ore mining methods.

			Layer			Stability of
Group		Subgroup	thickness	Dip angle	Ore stability	hanging wall /
			(m)	(0)		floor
	 Frontal ex 	xcavation methods	1,5-4	$\leq 30^{0}$		stable /
ZZ					stable	stable
DS	2. Room and	d pillar methods	2 - 30	$\leq 45^{\circ}$		stable /
Į Į Į					stable	stable
UN A EE	3. Sublevel	caving methods	≥30	$\geq 60^{0}$		stable /
A C B					stable	stable
E	4. Shrinkage	e stope methods	0,6-5	$\geq 60^{0}$		stable /
					stable	stable
		1.1. Roof excavations in	≥ 1	$\geq 60^{0}$		any/
	1. Excavation	horizontal levels			stable	any
	methods with	1.2. Roof excavations in sloping	0,7 - 4	$\geq 60^{0}$		stable /
S	backfilling of	levels			stable	stable
10	stopes	1.3. Self-backfilling excavation	0,1-0,8	$\geq 60^{0}$		any/
E B		methods			stable	any
E Z	2. Excavation	2.1. Excavation methods in	> 4	$\geq 50^0$		any/
	method with	horizontal levels			unstable	any
0 D	substructure and	2.2. Excavation methods by	> 4	0 - 900		any/
ATI H B	backfilling of	square set			unstable	any
NE	excavated areas			_		
M CA		3.1. Excavation methods in	>4	0 - 900		any
EX	3. Excavation	horizontal levels		_	unstable	
Π	methods with	3.2. Frontal excavation methods	> 3	$\leq 30^{0}$		moderately
	subdivision of				unstable	firm
	excavated areas	3.3. Excavation methods per dip	< 3	$\leq 45^{\circ}$		moderately
		of deposit		_	unstable	firm
Γτ.	1. Excavation meth	ods by the immediate roof caving	< 4	$< 40^{\circ}$	stable	moderately
Ο̈́ζ						firm
ALC DS	2. Excavation meth	od by roof caving in the levels	> 3	≥ 45	stable	unstable/
[O] [A]						unstable
	3. Sublevel caving	methods	>15	≥ 45	any	unstable/
ЯХË						unstable
E E ≥	4. Excavation meth	ods with block cutting	> 25	≥ 75	unstable	moderately
						firm

On the example of one deposit of irregular contours, with stable ore and unstable sides. The deposit has an average capacity of about 2 m, with dip angle of about 50° and a depth of about 300 m. The value of the ore in the deposit is average, and the distribution of useful minerals is uneven in such a way that the rich parts of the ore are mixed with the poor parts. Based on the known characteristics, Table 4. is compiled with a list of factors and their characteristics.

Table 4. Choice of the excavation method based on the natural characteristics of the deposit (Example)

Characteristic	Value	Possible method of excavation
Dip angle	50^{0}	I-3,4; II-1.1.,1.2.,1.3.,1.4. ,2,1.,2.2.,3.1,
		III-2,3,4
Deposit thickness	2 m	I-1,2,4;II-1.1,1.2,1.3,3.1,3.2;III-1,2
Strength		
1. Ore	Stable	II-1.1,1.3;III-2,3
2. Surrounding rocks	Unstable	
Deposit contours	Incorrect contour	
Contact of ore and surrounding rocks	Clear contact	I - III
	Rich ore alternates with poor	I - III
Metal distribution in the deposit	ore and waste	with selective excavation
Disturbance of the terrain surface	No limit	I - III
Depth of exploitation	300 m	I - III

Table 4. shows that the following excavation methods are suitable for a specific ore deposit:

- Method of roof directional excavation with backfilling in horizontal levels
- Sub-level caving method
- Excavation methods with backfilling and backfilling in horizontal levels
- Excavation methods with supporting in the horizontal levels

The final decision on the choice of excavation method will be made by optimizing of the selected excavation methods, which can be done on the basis of technical and economic parameters by economic comparison of variants of the excavation methods or by the method of multi-criteria optimization [7,8,9].

CHOICE OF EXCAVATION METHOD BY MULTICRITERIA ANALYSIS

The method of analytical hierarchical processes (AHP) was developed by Thomas Saaty in the early 1970s and it represents one tool in decision analysis. The area of method application is multi-criteria decision, where based on the defined set of criteria and attribute values for each alternative, the most acceptable solution is selected, i.e. the complete schedule of the importance of the alternative in the model is shown. Four phases of application of the method can be distinguished [10]:

- 1) structuring the problem
- 2) data collection
- 3) estimation of the relative weights
- 4) determining the solution to the problem

Problem structuring consists of decomposing a certain complex decision problem into a series of hierarchies, where each level represents a smaller number of the managed attributes. A graphical representation of the structuring of the problem is presented in Figure 1.



Figure 1. Structuring of the problem

By collecting data and measuring them, the second phase of the AHP method begins. The decision maker assigns the relative ratings to pairs of attributes of one hierarchical level, for all levels of the entire hierarchy. The Saaty scale of nine points, presented in Table 5, is used.

-		
Impotance	Definition	Explanation
intensity		
1	Equally important	It is used when both factors act equally on the final result.
3	Somewhat more	It is used when there is a small difference on the side of one factor in relation
	important	to another.
5	Much more	It is used when one factor is much more important than another factor.
	important	
7	Significantly more	It is used where one factor is significantly more important than another.
	important	
9	Absolutely more	It is used in the case when one instance is absolutely more important than
	important	another instance, without any doubt.
2,4,6,8	Intermediate values	They are used when a compromise between two values is needed. That is,
		when it is difficult to decide between two odd intensities of importance.

Table 5. Saaty scale

At the end of this phase, an appropriate matrix of comparisons by pairs corresponding to each level of the hierarchy is obtained.

Estimation of relative weights is the third phase of application of the AHP method. The comparison matrix will be "translated" in pairs into problems of determining of own values, in order to obtain normalized and unique own vectors, as well as weights for all attributes at each level of the hierarchy A_1

2, ..., A A_n , with the weight vector $m m m m = ()_{12}, ..., n_n, [10].$

Determining the solution of the problem is the last phase of the AHP method, and it involves finding the so-called composite normalized vector. Once the vector of the sequence of activity of the criteria in the model is determined, in the next round it is necessary, within each observed criterion, to determine the sequence of importance of the alternative in the model [11].

Finally, the overall synthesis of the problem is performed as follows: the participation of each alternative is multiplied by the weight of the certain criterion, and then these values are summed for each alternative separately. The obtained data represents the weight of the certain alternative in the model. In the same way, the weight is determined for all other alternatives, after which the final sequence of alternatives in the model can be determined.

The AHP method belongs to the group of popular methods, because it has the possibility of identifying and analyzing the consistency of the decision maker in the process of comparing elements from the hierarchy. Since the comparison of the alternative is based on a subjective assessment by the decision maker, it is necessary to constantly monitor it, in order to ensure the necessary accuracy [12]. The AHP method makes it possible to monitor the consistency of estimates at any time in the process of comparing the alternative pairs.

Using the consistency index:

$$C.I. = (\lambda_{max} - n)/(n - 1)$$

 $C.R. = \frac{C.I.}{R.I.}$ the consistency ratio is calculated: *R.I.*- random index (matrix consistency index of size *n* randomly generated pair comparisons).

n	1	2	3	4	5	6	7	8	9	10
R.I.	0	0	0,52	0,89	1,11	1,25	1,35	1,40	1,45	1,49

Table 6. Random index value R.I.

The coefficient λ_{Max} is the maximum and main characteristic of the value of the comparison matrix, while *n* is the size of the comparison matrix. In this case, it is valid that $\lambda \ge n$, and the difference $\lambda_{\text{max}} - n$ is used in measuring the consistency of the estimation. In case of inconsistency, if λ_{Max} is closer to *n*, the Majstorović, S. et al: Choice of Excavation Archives for Technical Sciences 2023, 28(1), 1-10

estimate is more consistent. If the comparison matrix is $C. R. \le 0,10$, estimates of the relative importance of the criteria (alternative priorities) are considered as acceptable. Otherwise, the reasons why the inconsistency of the assessment is unacceptably high need to be found.

The choice of the excavation method according to the natural characteristics of the deposit always precedes to the final choice, which means that in this way the variants of the excavation method are defined which according to the natural characteristics meet the set conditions [13]. For example, for the appropriate deposit the following excavation methods are selected that could be applied in Table 7.

Table 7. The variants of excavation methods selected according to the natural characteristics of the deposit

Serial number	Excavation method
1	Roof directional excavation method with backfilling in horizontal levels
2	Sublevel caving method
3	Methods of excavation with support and backfilling in horizontal levels
4	Excavation methods with support in horizontal levels

Attributes at the second level (decision criteria) are marked as follows (Table 8):

S.N.	Criterion	Mark
1	Value of excavated ore	$K_1 \rightarrow max$
2	Safety of excavation work	$K_2 \rightarrow max$
3	Processing coefficient (k _p)	$K_3 \rightarrow min$
4	Recovery of deposit (i)	$K_4 \rightarrow max$
5	Dilution of the obtained ore (r)	$K_5 \rightarrow min$
6	Production price of 1 t of ore (C _{pr})	$K_6 \rightarrow min$
7	Excavation effect (U _o)	$K_7 \rightarrow max$
8	Surface impact and other environmental impacts	$K_8 \rightarrow min$

Table 8. Criteria for choice of excavation method

Criteria such as coefficient of processing, recovery, dilution of ore, excavation effect for individual groups and subgroups of excavation methods of non-layered (ore) deposits are given in Table 9.

			Uo	k _p	i	r	C _{pr}
Group		Subgroup	t/shift	m/1000 t	%	(%)	(€/t)
PE	1. Frontal e	xcavation methods	30 - 70	0,5 - 3	70 - 90	5 - 15	17,5 – 25
STO HOD	2. Chamber	-pillar methods	30 - 70	2-14	60-80	5 - 15	20 - 30
PEN	3. Sublevel	caving methods	60 - 90	3 - 16	80 - 95	5 - 15	18 - 25
I O N	4. Shrinkag	e methods	25 - 40	2 - 17	75 - 90	3 - 15	25 - 32
	1. Excavation	1.1. Roof excavations in horizontal levels	10 - 25	3 - 9	95 - 99	1 - 5	27 - 40
SOOF	methods with backfilling of	1.2. Roof excavations in sloping levels	9 - 18	4 - 14	94 - 98	2 - 8	25 - 40
METI DING	stopes	1.3. Self-backfilling excavation methods	2 - 8	5 - 20	88 - 98	5 - 20	15 - 30
ATION H BUIL	2. Excavation method with support and	2.1. Methods of excavation in horizontal levels	5 - 20	2 - 12	85 - 95	5 - 15	20 - 40
CAV WIT	backfilling of stopes	2.2. Methods of excavation by square set	2 - 15	5 - 12	90 - 95	5 - 15	34 - 45

 Table 9. Influential techno-economic parameters for the choice of the excavation method

2 - 15

5 - 12

95 - 98

2 - 10

support

3.1. Excavation methods

in horizontal levels

II EXC

28 - 35

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	3. Excavation methods with	3.2. Frontal excavation methods	4 - 15	5 - 12	90 - 95	5 - 15	25 - 35
	subdivision of excavated areas	3.3. Excavation methods per dip of deposit	2 - 15	7 - 12	90 - 95	2 - 13	25 - 32
CAVING	1. Excavation met immediate hangin	hods by caving of the g wall	10 - 20	3 - 8	85 - 95	2 - 15	15 - 28
S WITH	2. Excavation met	hod by roof caving	15 - 35	4 - 6	95 - 98	1 - 8	12 - 30
ETHOD	3. Sublevel caving	g methods	40 - 90	3 - 15	80 - 95	5 - 25	15 - 25
M III	4. Excavation met	hods with block cutting	5 - 130	2 - 10	75 - 95	10 - 25	15 - 30

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The next step of the AHP method is pairwise comparison, Saaty's scale of relative importance is used when comparing in pairs.

	K ₁	К2	К3	K_4	K_5	К ₆	K ₇	K_8	Weights
К1	1,000	3,000	4,000	3,000	5,000	2,000	7,000	8,000	0,2792
К2	0,333	1,000	2,000	0,333	4,000	0,200	3,000	5,000	0,0967
К3	0,250	0,500	1,000	0,333	4,000	0,166	3,000	5,000	0,0799
K4	0,333	3,000	3,000	1,000	5,000	0,166	4,000	6,000	0,1417
K5	0,200	0,250	0,250	0,200	1,000	0,143	0,500	3,000	0,0359
K ₆	0,500	5,000	6,000	6,000	7,000	1,000	7,000	8,000	0,3016
К ₇	0,143	0,333	0,333	0,250	2,000	0,143	1,000	4,000	0,0454
K ₈	0,125	0,200	0,200	0,166	0,133	0,125	0,250	1,000	0,0195
Total	2,884	13,283	16,783	11,282	28,130	3,943	25,750	40,00	

Table 10. Comparison of attributes at the level of decision criteria

$$\lambda = 8,77$$
; $C.I. = \frac{\lambda_{max} - n}{n-1} = \frac{8,77-8}{7} = 0,11$; $C.R. = \frac{C.I.}{R.I.} = \frac{0,11}{1,4} = 0,079 < 0,10$

Analogous to the previous one, the attributes of the alternative level can be marked as follows (Table 11):

Table 11. Alternative level attributes

Roof directional excavation method with backfilling in horizontal levels	A ₁
Excavation method with roof caving	A ₂
Methods of excavation with support and backfilling in horizontal levels	A ₃
Excavation methods with support in horizontal levels	A_4

The corresponding alternative comparison matrices for each criterion attribute and their priorities are shown in the following tables (12 - 19).

Table 12. Matrices of relevant importance of the alternative in relation to attribute K_1 (Value of exc. ore)

	A1	A_2	A ₃	A_4	Weights
A_1	1,000	5,000	3,000	3,000	0,5050
A ₂	0,200	1,000	0,500	0,333	0,0868
A ₃	0,333	2,000	1,000	3,000	0,2441
A_4	0,333	3,000	0,333	1,000	0,1641
Σ	1,866	11,000	4,833	7,333	

$$\lambda = 4,25$$
; $C.I. = \frac{\lambda_{max} - n}{n-1} = \frac{4,25-4}{3} = 0,084$; $C.R. = \frac{C.I.}{R.I.} = \frac{0,084}{0,89} = 0,094 < 0,10$

	A1	A_2	A ₃	A4	Weights
A ₁	1,000	7,000	3,000	5,000	0,5769
A_2	0,143	1,000	0,333	0,500	0,0716
A ₃	0,333	3,000	1,000	3,000	0,2399
A_4	0,200	2,000	0,333	1,000	0,1125
Σ	1,676	13,000	4,666	9,500	

Table 13. Matrices of relevant importance of the alternative in relation to attribute K₂ (Work safety)

$$\lambda = 4,06$$
; C.I.= $\frac{\lambda_{max} - n}{n-1} = \frac{4,06-4}{3} = 0,02$; C. R. = $\frac{C.I.}{R.I.} = \frac{0,02}{0,89} = 0,022 < 0,10$

Table 14. Matrices of relevant importance of the alternative in relation to attribute K₃ (Processing coefficient)

	A_1	A_2	A ₃	A_4	Weights
A_1	1,000	5,000	3,000	2,000	0,4658
A_2	0,200	1,000	0,333	0,200	0,0691
A ₃	0,333	3,000	1,000	0,500	0,1679
A_4	0,500	5,000	2,000	1,000	0,2973
Σ	2,033	14,000	6,333	3,700	

$$\lambda = 4,058 \text{ ; C.I.} = \frac{\lambda_{max} - n}{n-1} = \frac{4,058 - 4}{3} = 0,019 \text{ ; C. R.} = \frac{\text{C.I.}}{\text{R.I.}} = \frac{0,019}{0,89} = 0,022 < 0,10$$

Table 15. Matrices of relevant importance of the alternative in relation to attribute K4 (Recovery of deposit)

	A_1	A_2	A_3	A_4	Weights
A ₁	1,000	7,000	5,000	4,000	0,6256
A_2	0,143	1,000	0,200	0,333	0,0614
A ₃	0,200	5,000	1,000	0,500	0,1659
A_4	0,250	3,000	2,000	1,000	0,1988
Σ	1,593	16,000	8,200	4,833	

$$\lambda = 4,236$$
; $C.I. = \frac{\lambda_{max} - n}{n-1} = \frac{4,236 - 4}{3} = 0,079$; $C.R. = \frac{C.I.}{R.I.} = \frac{0,079}{0,89} = 0,088 < 0,10$

Table 16. Matrices of relevant importance of the alternative in relation to the attribute K_5 (Dilution of the obtained ore)

	A_1	A_2	A_3	A_4	Weights
A ₁	1,000	0,143	0,333	0,200	0,0568
A_2	7,000	1,000	5,000	4,000	0,5917
A ₃	3,000	0,200	1,000	0,500	0,1302
A_4	5,000	0,250	2,000	1,000	0,2212
Σ	16,000	1,593	8,333	5,700	

$$\lambda = 4,124$$
; C.I.= $\frac{\lambda_{max} - n}{n-1} = \frac{4,124 - 4}{3} = 0,041$; C.R. = $\frac{C.I.}{R.I.} = \frac{0,041}{0,89} = 0,045 < 0,10$

Table 17. Matrices of relevant importance of the alternative in relation to attribute K_6 (Production price of 1t of ore)

	A_1	A_2	A_3	A_4	Weights
A_1	1,000	5,000	3,000	2,000	0,4583
A_2	0,200	1,000	0,333	0,200	0,0691
A ₃	0,333	3,000	1,000	2,000	0,2372
A_4	0,500	5,000	0,500	1,000	0,2247
Σ	2,033	14,000	4,833	5,200	

$$\lambda = 4,225 \text{ ; C.I.} = \frac{\lambda_{max} - n}{n - 1} = \frac{4,429 - 4}{3} = 0,074 \text{; C. R.} = \frac{\text{C.I.}}{\text{R.I.}} = \frac{0,075}{0,89} = 0,081 < 0,10$$

	A_1	A_2	A_3	A_4	Weights
A_1	1,000	0,200	3,000	3,000	0,2211
A_2	5,000	1,000	4,000	3,000	0,5479
A ₃	0,333	0,250	1,000	2,000	0,1308
A_4	0,333	0,333	0,500	1,000	0,1004
Σ	6,666	1,783	8,500	9,000	

Table 18. Matrices of relevant importance of the alternative in relation to attribute K₇ (Mining performance)

$$\lambda = 4,165$$
; C.I.= $\frac{\lambda_{max} - n}{n-1} = \frac{4,165 - 4}{3} = 0,055$; C.R. = $\frac{C.I.}{R.I.} = \frac{0,055}{0,89} = 0,062 < 0,10$

Table 19. Matrices of relevant importance of the alternative in relation to the attribute K₈ (Impact to surface)

	A1	A_2	A_3	A_4	Weights
A_1	1,000	0,250	2,000	3,000	0,2292
A_2	4,000	1,000	4,000	3,000	0,5101
A ₃	0,500	0,250	1,000	0,333	0,0919
A_4	0,333	0,333	3,000	1,000	0,6752
Σ	5,833	1,833	10,000	7,333	

$$\lambda = 4,225 \ ; \ C.I. = \frac{\lambda_{max} - n}{n-1} = \frac{4,225 - 4}{3} = 0,075; \ C.R. = \frac{C.I.}{R.I.} = \frac{0,075}{0,89} = 0,084 < 0,10$$

At the end of the procedure, a synthesis of the problem of choosing the excavation method is performed, so that all alternatives are multiplied by the weights of individual decision criteria, and the obtained results are added, as shown in Table 20.

Criterion	Weight	A1	Weight	A ₂	Weight	A ₃	Weight	A_4	Weight
	criteria		x A ₁		$x A_2$		x A ₃		x A ₄
К1	0,2792	0,5050	0,1409	0,0868	0,0242	0,2441	0,0682	0,1641	0,0458
К2	0,0967	0,5769	0,0558	0,0716	0,0069	0,2399	0,0232	0,1125	0,0109
К3	0,0799	0,4658	0,0372	0,0691	0,0055	0,1679	0,0134	0,2973	0,0237
К4	0,1417	0,6256	0,0886	0,0614	0,0087	0,1659	0,0235	0,1988	0,0282
К5	0,0359	0,0568	0,0200	0,5917	0,0212	0,1302	0,0047	0,2212	0,0079
K ₆	0,3016	0,4583	0,1382	0,0691	0,0208	0,2372	0,0715	0,2247	0,0678
К7	0,0454	0,2211	0,0100	0,5479	0,0249	0,1308	0,0059	0,1004	0,0046
K ₈	0,0195	0,2292	0,0044	0,5101	0,0099	0,0919	0,0018	0,6752	0,0132
			0,4774		0,1223		0,2122		0,2021

Table 20. Choice of the optimal alternative of the excavation method

The alternative with the highest value is the most acceptable or optimal method of excavation for a specific ore deposit, which means that alternatives A1 - Method of roof directional excavation with backfilling in horizontal levels will be selected for exploitation of this deposit.

CONCLUSION

The paper presents the procedure for choice of the optimal excavation method of ore deposit through two basic phase:

- 1. The choice of the excavation method according to the natural characteristics of the deposit.
- 2. The choice of the excavation method by the procedure of multicriteria analysis from the group of possible excavation methods selected on the basis of natural characteristics.

The choice of the excavation method by the procedure of multicriteria analysis was performed by applying the AHP method in specific conditions of the ore deposit. One of the leading problems in the application of this method is the definition of decision-making attributes at the second level (decision-making criteria) and the assessment of their relevant weights. The authors defined the criteria and estimated the values of their relative weight based on their own experiences in previous scientific research.

Received May 2022, accepted August 2022)

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