ANALYSIS OF BLASTING ROCKS PREDICTION AND ROCK FRAGMENTATION RESULTS USING SPLIT-DESKTOP SOFTWARE

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Abstract

One way to optimize the rock blasting operation is to conduct detailed structural analysis of the rock mass. For this, it makes use of algorithms that allows analyze the degree of fragmentation by changing the geometrical and geomechanical parameters, such as the equations proposed by Kuznetsov (1973), used for studying the interaction between explosive and rock. In this context, this study aimed to compare the classical methodology for the fragmentation of rocks with the method of fragmentation analysis by image. The study was done through field data provided by Herval Quarry which were analyzed for evaluation of fragmentation through images of blasted material muck piles. Were analyzed the P_{50} , P_{80} and P_{100} of granulometric distribution curves. The optimization of blasting was done by adjusting the mesh of the fire plan (burden and spacing) and stemming length.

Keywords: Kuz-Ram equations; Blasting; Rock fragmentation; Split-desktop.

ANÁLISE DA PREVISÃO DO DESMONTE DE ROCHAS E DOS RESULTADOS DA FRAGMENTAÇÃO UTILIZANDO O SOFTWARE SPLIT-DESKTOP

Resumo

Uma maneira de otimizar a operação de desmonte de rochas é a realização de uma análise estrutural detalhada do maciço rochoso. Para isso, faz-se uso de algoritmos que permitem analisar o grau de fragmentação, alterando os parâmetros geométricos e geomecânicos, tais como as equações propostas por Kuznetsov (1973), utilizadas para estudar a interação entre o explosivo e a rocha. Neste contexto, este estudo teve como objetivo comparar a metodologia clássica para a previsão da fragmentação de rochas com o método de análise de fragmentação por imagem. O estudo foi feito por meio de levantamento de campo realizado na Pedreira Herval para avaliação da fragmentação através de tratamento de imagens obtidas na pilha de material desmontado. Foram analisados o P_{50} , P_{80} e P_{100} das curvas de distribuição granulométrica. A otimização da detonação foi feita com o ajuste da malha do plano de fogo (afastamento e espaçamento) e o comprimento do tampão.

Palavras-chave: Equações de Kuz-Ram; Desmonte com explosivos; Fragmentação da rocha; Split-desktop.

I INTRODUCTION

Various authors have studied explosive-rock interaction models, which provide a useful description of the blasting process as a "background" process for modeling the fragmentation of the rock by explosives, such as Kuznetsov [1], Cunningham [2], Lilly [3], Sarma [4], and Djordjevic [5].

The rocks have geomechanical characteristics that depend on the geology of the rocks that form them and also some geotechnical parameters that are intrinsic of each mass and its tectonic history [6]. Rarely rock masses have a homogeneous structure, possessing a set of discontinuities that will have significant impacts on their geomechanical behavior [7,8]. Therefore, there is a need to previously determine the discontinuities in the rock mass and deforming stage reached in order to reduce rock blasting costs. An effective method in this process that will be object of study in this project is the evaluation of discontinuities, analyzing the deformation markers in rock masses, structural and petrographically for subsequent application of these discontinuities in the blast design through the mathematical model Kuz- Ram.

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I.I The Kuz-Ram Fragmentation Model

In this model, the properties of the rocks, the properties of the explosives and the geometric variables of blast plan are combined using five equations that make up the model of Kuz-Ram fragmentation [3]:

 Breaking theory: the amount of breakage that occurs with a known quantity of an explosive energy can be estimated using the Kuznetsov's Equation I [I].

$$X_{50} = A x \left(\frac{V_0}{Q_e}\right)^{0.8} x \left(Q_e\right)^{\frac{1}{6}} x \left(\frac{E}{115}\right)^{-\frac{19}{30}}$$
(1)

where X_{so} is the average particle size (cm), A is the rock factor, V_o is the volume of blasted rock by hole (m³), Q_e is the mass of the explosive used (kg) and represents the relative energy mass (RWS) of the explosive compared to ANFO [1].

 Theory size particle distribution (Equation 2): size distribution of the fragmented rock particles can be determined from the average size, the breakage model is known [9].

$$P = 100 x \left[1 - e^{-0.693 x \left(\frac{X}{X_{50}} \right)^n} \right]$$
 (2)

where P is the percentage passing, X is the mesh size sieve, and n is the uniformity index [9].

 Explosive Detonation Theory: the amount of energy released by explosive is calculated using Tidman's equation 3:

$$E = \left(\frac{VOD_e}{VOD_n}\right)^2 x RWS$$
(3)

where *E* is the effective power by relative mass of an explosive; VOD_e is the speed of effective explosive detonation (measured in the field); VOD_n is the nominal velocity of detonation of the explosive (m/s) and *RWS* is the energy per mass relative to ANFO (%) [10].

 Correlation of blast design parameters (Equation 4): there is a correlation between the various configurations of the blast plans and model of rock fragmentation [2]:

$$n = \left[2, 2 - 14x \left(\frac{B}{D}\right)\right] x \left[\frac{1 + \left(S / B\right)}{2}\right]^{0.5} x \left[\left(1 - \frac{W}{B}\right) x \left(\frac{L}{H}\right)\right]$$
(4)

where *B* is the burden (m); *S* is the spacing (m); *D* is the hole diameter (mm); *W* is the drilling standard deviation (m); *L* is the total length (m), and *H* is the bench height (m). When using two explosives in the hole (bottom load and column load), the equation is modified to Equation 5:

 $n = \left[2, 2 - 14x \left(\frac{B}{D}\right)\right] x$ $\left[\frac{1 + (S/B)}{2}\right]^{0.5} x \left\{ \begin{bmatrix} 1 - \frac{W}{B} \end{bmatrix} x \\ \begin{bmatrix} abs x \\ \left(\frac{BCL - CCL}{L}\right) + 0, 1 \end{bmatrix}^{0.1} x \left(\frac{L}{H}\right) \right\}$ (5)

where *BCL* is the length of the bottom charge (m); *CCL* is the length of the column load (m); *abs* is the absolute value concerning (*BCL* - *CCL*)/*L* [2].

 Rock type's correlation (Equation 6): the properties and characteristics of the rock mass interfere with the result of the fragmentation. The Lilly's geomechanical classification subsequently modified by Cunningham is used in Kuz-Ram fragmentation model [3]:

$$A = 0.06 x (RMD + RDI + HF)$$
(6)

where A is the rock factor, the values of *RMD* (Rock Mass Description) are obtained from the geomechanical classification of Lilly, and other parameters are obtained with the help of equations 7, 8 and 9:

$$RDI = 25d - 50$$
 (7)
 $HF = E / 3, se E < 50GPa$ (8)
 $HF = UCS / 5, se E > 50GPa$ (9)

where d is the density of the rock, E is Young's modulus (GPa), and UCS to the uniaxial compressive strength of the rock (MPa) [3].

The uniaxial compression strength was obtained through the equation proposed by Barton et al. [11] and modified by Gokceoglu et al. [12], which considers the classification of the rock mass by the Q system (Equation 10). As the rock mass was already classified according to the RMR system, a correlation was proposed by Goet et al. [13] between the RMR and Q systems to determine this system through the RCR index, which corresponds to RMR without the values referring to joint orientations and intact rock strength (Equation 11):

$$E = 10 x Q_c^{1/3}; Q_c = Q x (UCS / 100)$$
(10)

$$RCR = 8lnN + 30 \tag{11}$$

being N (number of rock mass) equivalent to the Q classification system considering the value of the Stress Reduction Factor (SRF) equal to 1.

Tecnol. Metal. Mater. Miner., São Paulo, v. 15, n. 1, p. 22-30, jan./mar. 2018

1.2 Fragmentation Analysis by Image Processing

For the analysis of rock fragmentation in blasting was used the Split-Desktop[®] Version 3.1 program and pictures got in the field. This program is a resource used in image processing to calculate the rock fragments PSD (Particle Size Distribution) through digital image analysis. Digital images can be acquired using a digital camera, and can come from the material in the square (muck pile), transport truck or conveyor belt.

2 METHODOLOGY

The methodology consisted basically in three stages. The first consisted in the recognition of the study area, which is the Herval Quarry, located in the city of Barreiros - PE, as well as the collection of technical data to determine the necessary parameters for geomechanical classification of rock mass; the second step consisted in the simulation of fragmentation through the Kuz-Ram model to obtain the particle size distribution curve; and at the third step was carried out the analysis of image processing for fragmentation using the Split-Desktop software.

The activities mentioned in the quarry for data collection were: recognition of mining fronts, fronts of measurement to determine the points of existing fractures, orientation of fractures attitudes (sense and diving) and detailed survey of the blast design. The Herval Quarry has two mining fronts and their extraction is done in benches, each mining front has three floors, subparallel and horizontal. After collecting the bench data provided by Herval Quarry, it was made a rock blasting, which was held on May 8, 2015.

To determine the RMR was necessary to identify the discontinuities presents on the bench face and measured the distances between them, as shown in Figure I, the presence

of water in the discontinuities, the weathering degree of the discontinuities and the characteristics of surface roughness. All information was measured in the field during the visit to Herval Quarry, as shown in Table 1.

The uniaxial compression strength was defined according to Equations 10 and 11. The RCR index value is 70; ergo, according to Equation 11, the value of N and consequently the value of Q is 148.41. Substituting in Equation 10, the UCS value found is 176.63 MPa. According to the Geomechanical classification table of Bieniawski [14], for this value of UCS, the RMR weight is 12 (Table 1).

2.1 Fragmentation simulation by Kuz-Ram model

This model requires information about the rock mass. In the equations given there are factors that concern about the geomechanical and geological characteristics of the rock. For this, the classification was made based on geomechanical classification proposed by Bieniawski [14]. For the simulation a spreadsheet was developed using Excel software to assist in the calculations of the equations proposed. The simulation results are presented in Tables 2 to 4 which show the blast design parameters, data from the Kuz-Ram model and rock fragmentation results of the Rosin-Rammler model to blasting.

Table 1. Rock mass classification of the Herval Quarry

Parameters	Weights
Uniaxial compressive strength	12
RQD (%)	20
Spacing between discontinuities	15
Condition of discontinuities	20
Water presence in discontinuities	15
TOTAL WEIGHT	82



Figure 1. Bench face showing the discontinuities and the distances between them.

Table 2. Monitoring the blasting at the Herval Quarry

Identification		Calculation of loads	
Quarry: Herval		Column load (m)	0.00
Bench:		Bottom load (m)	5.50
Date blasting: 19/06/2015		Full load (m)	5.50
		Load ratio (Kg/m³)	0.80
Hole diameter (pol)	3.0	Explosive consumed (kg)	
Burden (m)	1.6	Emulsion pumped	0.00
Spacing (m)	3.2	Cartridge emulsion 2 ^{1/4} "x24"	1650.00
Inclination (degrees)	15	Total	1650.00
Stem (m)	1.0		
Meters drilled	403.00	Explosive	Amounts
Total number of holes	62		
Average lenght of holes	6.50	Cartridge emulsion	1650.00
Blistering fator	1.5		
Average height	6.28	Binding shock tube (17/25/42) ms	61
Volume per hole (in situ)	33.28		
Total volume (in situ)	2063.36		
Total volume (swelling)	3095.00	Fuse	2
Explosive load	Diameter	Linear rate of loading (Kg/m)	
Pumped emulsion	3.00	4.69	
Cartridge emulsion	2.25	4.84	

The Figure 2 shows the theoretical particle size distribution of the blasting at Herval Quarry obtained by the Rosin-Rammler model.

2.2 Fragmentation Analysis by Image Processing

For analysis of the blasting fragmentation at the Herval Quarry by image processing, it was used the Split-Desktop[®] Version 3. I program and photos of the blasting results acquired in field stage. Split-Desktop program is a resource used in image processing to calculate the size distribution of rock fragments through digital image analysis in grayscale. Digital images can be acquired using digital camera or cellphone.

After image acquisition, analysis of fragmentation from the blasting is carried out in five stages: opening of the images by the program; delineation of images; manual delineation edition to minimize errors; determination of images scale; analysis of the particle size of each image and blasting.

The acquisition and scheduling of the images were obtained with the aid of two spherical objects that were positioned apart in the lower and upper part of the image, so that their diameters were perpendicular to the optical axis of the image for that it did not differ from the real size.

Were captured 5 images of the blasting, which were delineated by the Split-Desktop. After processing the blasting pictures it was combined the graphs through the own Split-Desktop and generated a single graph for analysis of fragmentation. Figure 3 shows an example of the images captured for analysis of fragmentation along with their delineation by software, and Figure 4 shows the size distribution curve obtained by combination of all graphs of blasting rocks analysis by image processing.

3 RESULTS AND DISCUSSIONS

To compare the size distribution curves of the simulations and results of imaging were analyzed three basic points of the curves:

- P₅₀ The diameter size through which 50% pass (fifty percent of the particles);
- P₈₀ The diameter size through which 80% pass (eighty percent of the particles);
- P₁₀₀ The maximum diameter through which 100% pass (one hundred percent of the particles).

The evaluation of these curves showed that the Kuz-Ram model features a higher percentage of fines that the analysis processing of image curve. The two curves are similar in the range of about 20 cm to about 50 cm; above 50 cm the Kuz-Ram model has more fines. This can be demonstrated by plotting the two curves in the same simulation screen, as shown in Figure 5.

With regard to the P80, it may be noted that there is a small difference. This parameter obtained the value of approximately 60 cm for the simulation and the value of about 72.50 cm for size analysis by image. The discrepancy with the value by image analysis is approximately 17.24%.

Table 3. Kuz-Ram model data from rock blasting

		Description	Classification	Index
RMD	Description of the rock mass		Friable	10
			Fractured	JF
			Massive	50
JF		Fractured massive	JPS + JPA	
	Spaci	ing of discontinuities (m)	< 0.10 m	10
JPS			0.10 a MS	20
			MS - DP	50
MS		Oversize primary	[,] crushing (m)	
DP	Drilling loop parameters (m)			
	Direction ar	nd dip with respect to free face	Horizontal	10
IDA			Dipped out the free face	20
JFA			Direction perpendicular to the free face	30
			Plunged into the free face	40
RDI	Inf	luence density (g/cm³)	RDI = 25d - 50	18.75
UE		If E < 50 GPa	HF = E/3	0
HF	If E > 50 GPa		HF = UCS/5	35.33
Е	Yc	oung's Modulus (GPa)		64
USC	Uniaxial	compressive strength (MPa)		176.63
	Rock factor (A)		Tidman equation: Energy Explo	osive (Er)
RMD		50	VOD	4100
RDI	18.75		VODn	5000
HF		35.33	RWS	86
Α		6.2448	Er	57.8264
		Kuznetsov equati	on	
V _o		Rock blasting volum	ne	2063.36
Qe		Explosive mass (Kg	<u>z)</u>	1650.44
К		Load ratio (Kg/m³)	0.80
X ₅₀		Average size of the partie	cle (cm)	39.76
		Cunningham Uniformity	/ Index (n)	
	В	Burden (m)	1.6	
	S	Spacing (m)	3.2	
	D	Hole diameter (mm)	76.2	
	W	Standard deviation drilling (m	.) 0.00	
	L	Total length of the load (m)	5.50	
	н	Bench height (m)	6.50	
	N		1.975	

In relation to P50, there is a small discrepancy in relation to the simulated curve obtained by imaging of about 5.67%, having a diameter of 42.15 centimeters in image analysis and 39.76 cm for the simulation through Kuz-Ram.

It may be noted that the slope in curves appear to be the same, since it is the same blast plan; another common characteristic between the two curves is that they are in different size ranges; and a last observable characteristic is that the curve for the image processing analysis does not show results for sizes smaller than 10 cm, unlike the curve referring to the Kuz-Ram model, which provides sizes in the range of 0 to 10 cm.

In order to optimize the blast, the particle size was adjusted so that P80 had value approximately less than or equal to 50 cm, which corresponds to the primary crusher feeding, which is a jaw crusher model FAÇO Of 80x50 cm.

For this, an optimized blast design was developed, adjusting the drilling grid and the stem, based on the simulation

Table 4. Data of rock fragmentation in the blasting

Rosin-Rammler equation (P)			
Rock Fragmentation			
	Sieve (cm)	Passing (%)	
	I	0.048008388	
x	2	0.188636898	
	3	0.419711810	
	4	0.739674470	
	5	1.147023734	
	10	4.434833923	
	50	66.37139288	
	100	98.62275220	
	150	99.99284590	
	160	99.99804600	
	170	99.99950810	
	180	99.99988590	
	190	99.99997560	
	200	99.99999520	

of the Kuz-Ram model that has already been performed (Table 3). In relation to the drilling grid, there was a change in spacing, which decreased from 3.2 meters to 3 m, and in the burden, which decreased from 1.6 to 1.5 m; the stem, which was 1.0 m, decreased to 50 cm. With these changes, the number of holes increased from 62 to 71, increasing the cost of drilling and with explosives, but reducing with crushing and secondary blasting.

The results of optimization of blast design parameters are shown in Table 5.

The graph of Figure 6 shows the particle size curve of the optimized blast through the Rosin-Rammler equation:

It can be seen that through the small changes in the geometry of the fire plane it was possible to increase the fragmentation and obtain a P80 of approximately 50 cm, which optimizes the crushing operation and will probably cause a reduction in production costs.



Figure 2. Particle size curve simulation of rock blasting.



Figure 3. Image captured for analysis and their delineation by Split-Desktop program.

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Figure 4. General graph of particle size analysis of rock blasting for image processing.



Figure 5. Kuz-Ram model x Split-Desktop software to rocks blasting.

Table 5. Blast design optimized

Identification		Calculation of loads	
Quarry: Herval		Column load (m)	0.00
Bench:		Bottom load (m)	6.00
Date blasting: 19/06/2015		Full load (m)	6.00
		Load ratio (Kg/m³)	0.99
Hole diameter (pol)	3.0	Explosive consum	ied (kg)
Burden (m)	1.5	Emulsion pumped	0.00
Spacing (m)	3.0	Cartridge emulsion 2 ^{1/4} "x24"	2061.84
Inclination (degrees)	15	Total	2061.84
Stem (m)	0.5		
Meters drilled	461.50	Explosive	Amounts
Total number of holes	71		
Average lenght of holes	6.50	Cartridge emulsion	2061.84
Blistering fator	1.5		
Average height	6.28	Binding shock tube (17/25/42) ms	71
Volume per hole (in situ)	29.25		
Total volume (in situ)	2076.75		
Total volume (swelling)	3115.10	Fuse	2
Explosive load	Diameter	Linear rate of loading (Kg/m)	
Pumped emulsion	3.00	4.69	
Cartridge emulsion	2.25	4.84	



Figure 6. Particle size curve simulation of optimized rock blasting.

4 CONCLUSIONS

After performing the particle size analysis by digital image processing with the Split-Desktop program, it was observed that the rock mass characteristics are essential to the best performance of the blasting operation. A more detailed study of rock mass, beside reduce transportation, handling and crushing costs, reduce the cycle time by optimizing these operations and increasing production.

The image processing analysis played a fundamental role to assess the relationship between the blasting carried out and mathematical modeling of the rock mass, because it was possible to compare the result of the particle size distribution generated by simulation through the Kuz-Ram model, with the practical result generated by blasting.

It can be seen from simulation of blasting at Herval Quarry that the increase of-drilling grid contributes to blast results with large particle sizes, although reducing the amount of explosive in bench (powder factor). Other operational inconveniences are: increased wear of crushing and loading equipment, higher maintenance time, and higher fuel consumption. It can be said that the Kuz-Ram fragmentation model assisted by simulation shows efficiency in predicting the rock fragmentation using explosives. Furthermore, the model has limitations, such as:

- Overestimation of the amount of fines;
- The model doesn't predicts the outcome of the particle size of rock fragmentation for different types of mesh, and it is known that the type of drilling grid applied influence the fragmentation blasting;
- Another important factor that must be considered is that there is no prediction or modeling to analyze the tie-in sequence, the maximum load expected nor influences the use of different delay time.

Despite the limitations of both the model and the simulator, this tool proves important in predicting blasting aiming to obtain a suitable drilling grid in order to maintain a normal operating level and the desired degree of fragmentation by blasting.

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Recebido em: 26 Set. 2016 Aceito em: 13 Maio 2017