



**NAMIBIA UNIVERSITY  
OF SCIENCE AND TECHNOLOGY**

**FACULTY OF ENGINEERING AND THE BUILT ENVIRONMENT**

**DEPARTMENT OF MINING AND PROCESS ENGINEERING**

<b>QUALIFICATION : BACHELOR OF ENGINEERING IN MINING ENGINEERING</b>	
<b>QUALIFICATION CODE: 08BMIN</b>	<b>LEVEL: 7</b>
<b>COURSE CODE: DBS721S</b>	<b>COURSE NAME: DRILLING AND BLASTING 323</b>
<b>SESSION: NOV 2022</b>	<b>PAPER: THEORY</b>
<b>DURATION: 3 HOURS</b>	<b>MARKS: 100</b>

<b>FIRST OPPORTUNITY EXAM PAPER</b>	
<b>EXAMINER(S)</b>	<b>Mr L. Madziwa</b>
<b>MODERATOR:</b>	<b>Mr P. Shava</b>

<b>INSTRUCTIONS</b>
<ol style="list-style-type: none"><li>1. Answer all questions.</li><li>2. Read all the questions carefully before answering.</li><li>3. Marks for each question are indicated at the end of each question.</li><li>4. Please ensure that your writing is legible, neat and presentable.</li></ol>

**PERMISSIBLE MATERIALS**

1. Examination paper
2. Scientific calculator

**THIS QUESTION PAPER CONSISTS OF 6 PAGES** (Including this front page)



**Question 1 [20 marks]**

- a) What are the effects of the delay timing and the initiation sequence of a blasting are not adequate? [4]
- b) Blast designs are undertaken using controllable and uncontrollable factors. Describe how five key controllable factors and five uncontrollable factors affect blast designs. [10]
- c) Explain how drillhole diameter, burden and powder factors variables affect fragmentation. [6]

**Question 2 [12 marks]**

A blast round was designed in a way that the diameters of the drilled holes were 50 mm, 75mm and 115mm with spacing to burden ratio 1:2. The bench height was determined by the length of the drill rods coupled together which are 3 meters each. 0.45 m deviation was recorded after using four of those drill rods. If rock constant is 10, and the density of the ANFO is 900kg/m<sup>3</sup>.

- a) Determine the loaded length, quantity of explosives and powder factor required for the three-hole diameters, taken the mean fragment size to be 30 cm, L is the loaded length above the toe, J is the hole stemming, H is the bench height, D is the diameter of drilled hole and  $S_{ANFO}$  is relative specific weight of ammonium fuel oil. Calculate the loaded hole length of explosive, the quantity of explosive as well as powder factor? Each parameter carries one mark.

Hole diameter, D (mm)	Charge Length, L (m)	Amount explosives, Q <sub>e</sub> (Kg/hole)	Powder factor, K (Kg/m <sup>3</sup> )
50	.....	.....	.....
75	.....	.....	.....
115	.....	.....	.....

- b) Draw your conclusion from the answers in (a) [3]



**Question 3 [14 marks]**

The open pit mine has massive fine-grained lava with a skarn gold mineralization and there is little jointing closer than anticipated drilling pattern. The mine management has requested you as a mining engineer to determine the rock factor. The massive grained lava rock has the following data:

UCS = 500MPa,                      Young's Modulus = 80 GPa  
Density = 2.8 t/m<sup>3</sup>,                      RDM =50,                      J=0

Determine the following;

- a) Rock density influence (RDI) [2]
- b) Hardness factor (HF), [2]
- c) Rock Factor (A) [3]

**Question 4 [20 marks]**

In blasting operations, drilling, and how explosives are charged, and the quantity of explosives used are major contributors to the output of a blast.

- a) Outline the effects of fragmentation on drilling, blasting, loading and hauling cost? [4]
- b) Explain what you understand by the term initiation system in blasting. [2]
- c) Explain what you understand by the term initiation sequence in blasting. [2]
- d) Why is hole orientation important in drilling? [2]
- e) Define cushion blasting using open pit mining and underground mining example. [4]
- f) What is smooth wall blasting? [2]
- g) What is pre-splitting? [2]
- h) What is post-splitting? [2]

**Question 5 [14 marks]**



If a mine drift is to be excavated in rock ( $c = 0.4$ ) by means of blasts with parallel blast holes and four sections cut, knowing that the geometric dimensions and drilling data are:

- Tunnel width  $AT = 4.5$  m
- Abutment height = 4.0m
- Height of arch = 0.5 m
- Relief hole diameter  $D_2 = 102$ mm
- Drilling diameter  $D_1 = 45$ mm

The explosive to be used has a Relative Weight Strength with respect to ANFO of 1.09 (109%) and the available cartridges have diameter of 25, 32 and 38mm, which give lineal charge concentrations for density of 1.2 g/cm<sup>3</sup>, of 0.59, 0.97, and 1.36 kg/m respectively.

- a) You require to determine the advance per round blasted, the development tunnel is about 60m, how many advances do you need to complete this tunnel? [6]
- b) Calculate your first, second, third and fourth section quadrangle cut? [8]

#### **Question 6 [20 marks]**

B&E Mining and Development contract has been awarded a contract for underground tunnel development. The contractor has requested you as mining engineer to determine certain parameters for their underground tunnel development. Some of the known parameters are as follows:

- Tunnel height = 6.5 m
- Targeted powder factor = 1.7 kg/m<sup>3</sup>
- 28 perimeter holes loaded with 0.5 kg load.
- The chosen burn cut requires 7 loaded holes with 2.8 kg load.
- Blasthole load weight = 5.5 kg/hole
- The round length = 3.4m

Calculate the following;

- a) Round volume (V), [4]
- b) Total Explosive weight (TW), [4]
- c) Blasthole weight (BW), [4]
- d) The cut weight (CT), and [4]





e) The number of blastholes (N).

[4]

**End of Examination**



**Important formulae for DLB 721S – 2022**

1. 
$$K_{act} = \frac{(L + U) * M_c}{B * S * H}$$

2. 
$$B = 0.012 \left( \frac{2SG_E}{SG_R} + 1.5 \right) \times D_E$$

3. 
$$M_c = \frac{C * \rho * d^2}{100 * 1273}$$

4. 
$$B = \sqrt{\frac{LM_c}{aHK_{tech}}}$$

5. 
$$V = \frac{\pi D^2}{4} x RD$$

6. 
$$Q = \frac{\pi d}{4 * 1000} \rho$$

7. 
$$P = 5.7 \times 10^{-5} \times (RF - 28 \times \log(0.145 \times S_c)) \frac{W}{\phi} \times RPM$$

$$P = (61 - 28 \log_{10} S_c) \frac{W}{\phi} \bullet \frac{rpm}{300}$$

7.a.

$$Um = 264p^{1/2}d^{1/2}$$

Where: d = diameter of the chip in inches  
p = density of chip in Ib/ft3

8.

9. 
$$VM = 2 * VP^{0.65} = 2 * 50^{0.65} =$$

10. 
$$RDI = 25 * RD - 50 = 25 * 2.8 - 50 = 20$$

11. 
$$HF = UCS / 5 = 500 / 5 = 100$$

12. 
$$\text{Rock Factor (A)} = 0.06(RMD + JF + RDI + HF) = 0.06(50 + 20 + 100) = 10.2$$

13. 
$$P = 2.5 \times pv^2 \times 10^{-6}$$

