

# Experimental investigation and numerical simulation of chip formation mechanisms in cutting rock-like materials

Aresh, B., Nawaz Khan, F. & Haider, J.

Author post-print (accepted) deposited by Coventry University's Repository

## Original citation & hyperlink:

Aresh, B, Nawaz Khan, F & Haider, J 2022, 'Experimental investigation and numerical simulation of chip formation mechanisms in cutting rock-like materials', Journal of Petroleum Science and Engineering, vol. 209, 109869.

<https://doi.org/10.1016/j.petrol.2021.109869>

DOI 10.1016/j.petrol.2021.109869

ISSN 0920-4105

Publisher: Elsevier

© 2022, Elsevier. Licensed under the Creative Commons Attribution-NonCommercial-NoDerivatives 4.0 International

<http://creativecommons.org/licenses/by-nc-nd/4.0/>

Copyright © and Moral Rights are retained by the author(s) and/ or other copyright owners. A copy can be downloaded for personal non-commercial research or study, without prior permission or charge. This item cannot be reproduced or quoted extensively from without first obtaining permission in writing from the copyright holder(s). The content must not be changed in any way or sold commercially in any format or medium without the formal permission of the copyright holders.

This document is the author's post-print version, incorporating any revisions agreed during the peer-review process. Some differences between the published version and this version may remain and you are advised to consult the published version if you wish to cite from it.

1       **Experimental Investigation and Numerical Simulation of Chip**  
2       **Formation Mechanisms in Cutting Rock-like Materials**

3                   **Balaji Aresh<sup>1,\*</sup>, Fahd N Khan<sup>2</sup>, Julfikar Haider<sup>3</sup>**

4       <sup>1</sup>*Faculty of Engineering, Environment, and Computing, Coventry University, Coventry, UK*

5       <sup>2</sup>*Faculty of Materials and Chemical Engineering, Department of Materials Science, Ghulam*  
6       *Ishaq Khan Institute of Engineering Sciences and Technology, Topi 23640, Pakistan*

7       <sup>3</sup>*Advanced Materials and Surface Engineering (AMSE) Research Centre, Manchester*  
8       *Metropolitan University, Chester Street, M1 5GD, UK*

9       **\*Corresponding author.**

10      Dr Balaji Aresh

11      Faculty of Engineering, Environment, and Computing,

12      Coventry University,

13      Coventry, UK

14      Tel: +44-7702-875841

15      Email: ad5851@coventry.ac.uk

16

17 **Abstract**

18 In this study, the effects of tool geometry such as rake angle, and cutting parameters such as  
19 depth of cut on the cutting forces were studied and correlated with the built-up edge during the  
20 material removal process of a rock-like workpiece. Cutting or scratch tests were performed on  
21 low and high strength simulated rock-like materials using a tungsten carbide tipped orthogonal  
22 drag tool with three different rake angles ( $0^\circ$ ,  $10^\circ$  and  $20^\circ$ ) in a custom-made machining set-up  
23 incorporating a high-speed video camera. Force data were measured by a tri-axial  
24 dynamometer and a compatible data acquisition system, and specific cutting energy was  
25 calculated to assess the material removal performance. Experiments showed that a cutting tool  
26 with a  $20^\circ$  rake angle produced an efficient cut. The high-speed video at the cutting edge were  
27 analysed to comprehend the formation and growth of the built-up edge. Novel insight was  
28 gained by characterising the shape and was observed that the constantly evolving shape was  
29 unique to each rake angle used, this creates an apparent rake angle. By varying the rake angle  
30 and cutting parameter, the measured cutting force and thrust force showed that the material  
31 strength, cutting tool geometry and depth of cut played important roles in removing materials.  
32 Higher cutting efficiency was indicated by lower specific cutting energy at higher depth of cut  
33 for all cutting conditions. The formation of the crushed zone in relation to the cutting force  
34 revealed that the cutting force increased with the size of the crushed zone having two types of  
35 chip formation modes: shearing and fracturing. Numerical simulations were performed using  
36 a commercially available tool called ELFEN, a hybrid finite-discrete element software  
37 package. The simulations correlated well with experimental investigation. The simulations  
38 also showed the formation of crushed zone and crack growth as observed experimentally  
39 through the use of high-speed video and also shed light on the state of stress state at the cutting  
40 edge.

41 **Keywords**

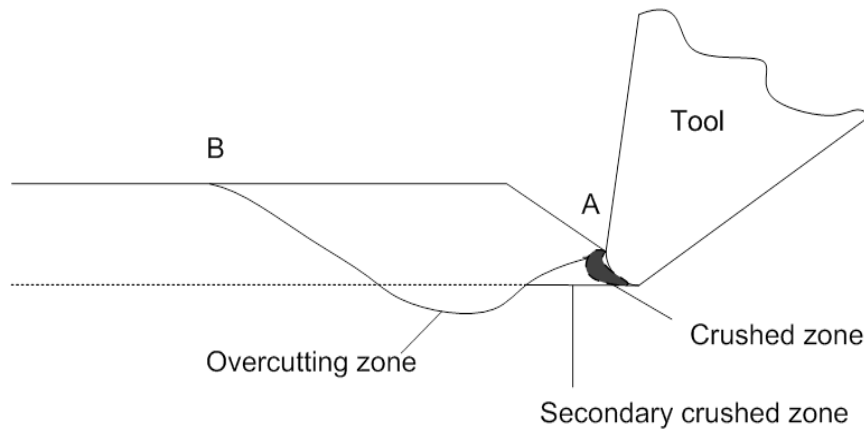
42 Fracture mechanics, Simulated rock, Crushed zone, Specific cutting energy, Rake angle, High  
43 speed video, Numerical modelling.

44

45 **1. Introduction**

46 Rock drilling is not only associated with the search for natural resources such as oil, gas and  
47 geothermal energies but also the activities related to collection of geological materials (e.g.,  
48 cores and cuttings) for scientific research, space exploration, data logging, search for water and  
49 mining and civil engineering (Rostamsowlat et al., 2022). Rock drilling tools have evolved  
50 over the years, undergoing changes in their design and in the materials used for their  
51 manufacturing. The technical feasibility and economic viability of processing of rocks (drilling,  
52 cutting, crushing, etc.) and the choice of a rock machining tools are dependant, inter alia, on  
53 the rocks.

54 Modern day rock drill bits are equipped with steel cylinders containing cemented carbide or  
55 Polycrystalline Diamond Compacts (PDC) teeth or milled steel with tungsten carbide coating  
56 or button inserts (Rostamsowlat et al., 2018). Improvement in the drilling and cutting of rocks  
57 needs a better understanding of the breakage and disintegration of the polycrystalline materials.  
58 The chip formation occurs in rocks due to elastic brittle deformation (Nishimatsu, 1972; O.D  
59 and L, 1974). Figure 1 shows the process of chip formation. As the tool pushes into the  
60 workpiece, the material ahead of the extreme cutting edge is crushed into a fine powder, this is  
61 known as the ‘crushed zone’ analogous to the built-up edge in metal cutting. The compaction  
62 of the crushed zone leads to the formation of a secondary crushed zone, which initiates the  
63 formation of a crack. The crack then propagates downwards, extending below the depth of cut  
64 and later rapidly rises up to the free surface resulting in a sudden fracture and the formation of  
65 a chip. The tool then moves on to meet a fresh face of the workpiece and the process repeats.  
66 The crack patterns are complex to understand given the anisotropy present in the brittle  
67 materials like rocks and they vary according to the cutting tools used (A. W, 2004; Cook et al.,  
68 1984; Dong, 1993; O.D and L, 1974; Suwannapinij, 1975). The study of the state of stress will  
69 provide a better understanding of the crack patterns, crack initiation and its propagation. The  
70 crushed zone influences the fracturing of rocks; though the fracture mechanism itself is elastic  
71 brittle deformation, the zone remains inelastic (Lindqvist and Hai-Hui, 1983) and since it  
72 propagates the energy from the tool to the rock, the importance of this zone cannot be  
73 overlooked.



74

75 *Figure 1. Chip formation model in rock cutting (adapted from [1])*

76 Specific Cutting Energy (SCE) is the energy consumed to remove a unit volume of material.  
 77 This parameter has been successfully used to characterise the efficiency of a cutting tool in  
 78 metal cutting (Khan et al., 2014, 2012; Sarwar, 1998; Sarwar et al., 2009, 2007) and it has been  
 79 used in the drilling industry as a measure of cost per unit volume of rock removed (Atici and  
 80 Ersoy, 2009) and correlated with other rock properties (Ersoy and Waller, 1995; Mohammadi  
 81 et al., 2020; Rostamsowlat, 2018; Wang et al., 2018; Yurdakul and Akdasç, 2012).

82 The equation for calculating the specific cutting energy is given by Equation 1:

$$E_{sp} = \frac{F_c L}{V} \quad (1)$$

83 where  $E_{sp}$  is the specific cutting energy,  $F_c$  is the mean cutting force,  $L$  is the length of cut, and  
 84  $V$  is the volume of rock removed.

85 Atici and Ersoy (2009) studied the specific energy for sawing and drilling of rocks, derived  
 86 from the energy required to remove a given volume of rock. Low values of SCE indicate  
 87 efficient cutting. Wang et al. (2018) conducted detailed rock cutting tests, and showed that SCE  
 88 was not only a function of rock properties but it was closely related to operational parameters  
 89 such as depth of cut and tool geometry. Huang and Wang (1997) investigated the process of  
 90 coring of rocks using diamond impregnated drill bits and found a correlation between weight-  
 91 on-bit (WOB) and SCE. The influence of tool geometry and depth of cut on the drilling of  
 92 rocks have been found in good correlation in a study conducted by Copur (2010). Ersoy and  
 93 Waller (1995) studied the relationship between WOB, rate of penetration and SCE and found

94 that as the WOB increases, so does the penetration rate, with a decrease in SE until an optimum  
95 WOB is reached.

96 The application of SE as performance indicators for roadheaders and tunnel boring machines  
97 have been researched by Acaroglu et al. (2008). Cho et al., (2010) used the SCE calculated  
98 from numerical simulation to derive the optimum spacing for TBM disc cutters. The influence  
99 of depth of cut was studied by Mohammadi et al. (2020) by conducting cutting tests using a  
100 chisel shaped cutting tool on various rock types and were able to correlate the debris size to the  
101 SCE. SCE was used to develop a new rippability classification system, as it can be easily  
102 determined without detailed on-site testing (Basarir et al., 2008).

103 SCE is often correlated with other rock parameters and efficiency indicators to provide  
104 operators with optimum running conditions for all types of rocks. Roxborough (1987) found  
105 an increase in SCE as the compressive strength of rocks increase. Coarseness Index (CI) is a  
106 comparative size distribution of the rocks. Tuncdemir et al. (2008) successfully correlated CI  
107 to SE and formed a statistical relationship defined by Equation 2.

$$SE = \frac{k}{CIn} \quad (2)$$

108 where k is function of rock strength and cutting tool parameters and n varies from 1.2 to 4.4  
109 based on the cutting tool.

110 Sengun and Altindag (2013) correlated SCE and the mechanical properties of rocks and found  
111 a strong correlation between the density, compressive strength and porosity. Atici and Ersoy  
112 (2009) found significant statistical correlation between SCE and the brittleness of rocks.  
113 Tiryaki and Dikmen (2006) found positive correlation between SCE and the textural and  
114 compositional properties of rocks.

115 Yadav et al. (2018) conducted cutting experiments on a model soft rock, Gypsum. Through a  
116 series of experiments using orthogonal cutting tools, the impact of positive and negative rake  
117 angles on the failure mechanism at the cutting edge was assessed. It was identified that for a  
118 positive rake angle, the failure mechanism was dominated by fracture and with a negative rake  
119 angle, a ductile type failure occurred. High speed camera and image analysis also highlighted  
120 a 'dead zone' forming at the cutting edge for a negative rake angle cutting tool where  
121 compacted particles of gypsum would adhere to the cutting edge. Physical models have long

122 been employed by engineers to provide qualitative as well as quantitative data (Stimpson,  
123 1970). Qualitative data is provided by the models which maintain geometric similarity, for  
124 example, scaled down models of buildings. This study deals with the physical models of the  
125 quantitative type where conditions of ‘similitude’ are maintained. Similitude can be achieved  
126 by simulating the physical and mechanical properties of rocks such as their brittle nature,  
127 compressive strength and elastic modulus, and as in the case with any other modelling work, it  
128 cannot be exact but a fair degree of accuracy can be maintained. The model materials as  
129 opposed to the actual rock samples have the advantage of being economically viable to produce  
130 or acquire within a short period of time, and being able to change the mechanical properties.  
131 Some examples of modelling materials are Portland cement, sand, plaster of Paris and dental  
132 plaster to name just a few. Materials are either classified as granular (e.g., sand, chalk, sawdust)  
133 or non-granular (e.g., glass, resin, ice), each having distinctive advantages and disadvantages  
134 over each other. The ease of sample preparation and time are some of the various factors which  
135 influence the choice of the model materials. Tien et al. (2006) used cement and kaolinite to  
136 simulate transversely isotropic rock, whereas plaster of Paris was used by Ozbay et al. (1996)  
137 to study the fracture process in highly stressed rocks. Sulfaset synthetic rock was used to study  
138 the shear stress test by Cho et al. (2008) and a mixture of barite, sand and plaster was used to  
139 study crack coalescence by Wong and Chau (1998).

140 Finite Element Methods (FEM) is a most common numerical method used to solve for a variety  
141 of engineering problems, but since it is based on the continuum concept, when applied to rock  
142 fracture mechanics, FEM fails to provide useful information when elements are required to  
143 open and separate (Jing and Hudson, 2002). Discrete Element Methods (DEM) works on the  
144 principle that the system is made up of both rigid and deformable bodies and when  
145 deformation/separation occurs then contact between the bodies are continually updated to  
146 ensure crack initiation and propagation, this however results in an increase in computational  
147 cost (Jing, 2003). DEM has been applied in a variety of problems from soil tilling to rock  
148 cutting. Ucgul et al. (2018) applied EDEM a DEM software tool to study the interaction  
149 between soil and tillage cutting tool. The cutting performance of conical picks influenced by  
150 rock brittleness was studied by Xuefeng et al. (2018) using a DEM software with Particle Flow  
151 Code in two dimensions (PFC<sup>2D</sup>).

152 Combined Finite element method (FEM) and discrete element method (DEM) is a powerful  
153 method to analyse the large number of fractures, since the continuum state of the rock mass

154 changes to a discontinuum state when the cracks initiate and propagate; this change of state is  
155 accomplished by a coupled FEM/DEM method. Carpinteri et al. (2004) used FRANC2D  
156 software developed by Cornell University to simulate rock indentation and ploughing on  
157 heterogeneous material using a discrete model and homogenous material using a FEM model.  
158 They observed stress patterns which indicate tensile parting of cracks and plastic crushing. Cai  
159 and Kaiser (2004) successfully used the ELFEN software to simulate the Brazilian Tensile  
160 Strength test on homogeneous rock, layered rock and rocks with pre-existing cracks. Li et al.  
161 (2021) successfully applied FEM/DEM method to simulate and study crack growth and how it  
162 is influenced by rake angle, back rake angle and depth of cut of a PDC cutter. The ELFEN  
163 software integrates FEM/DEM to provide a seamless change from continuum state to a  
164 discontinuum state.

165 Although some studies offered a certain degree of understanding on chip formation mechanism,  
166 but the scientific knowledge on the interaction between the tool and rock at the extreme cutting  
167 edge especially at the microscale level is still lacking. To fill this gap, this research will produce  
168 new knowledge of the rock cutting/deformation action at the extreme cutting edge using single  
169 cutting tooth test rig, high speed photography to study the built-up edge at the tool tip with  
170 synthetic rocks and the use of a coupled discrete/finite element code. The contributions in this  
171 work involves preparing artificial rock type materials with specific mechanical properties,  
172 develop an experimental set-up with associated instrumentation for cutting force measurement,  
173 understanding the chip formation mechanisms at the tool tip with a high speed camera system  
174 and conducting extensive experimental work to develop empirical models representing  
175 complex relationships among the tool geometry, workpiece materials and depth of cut with the  
176 specific cutting energy. Furthermore, numerical simulations were carried out to simulate the  
177 material removal process in rock cutting in line with the experimental conditions.

## 178 **2. Materials and Experimental Methods**

### 179 ***2.1. Workpiece specimen preparation***

180 Rock-like workpiece samples were prepared to simulate low strength (LS) and high strength  
181 (HS) rocks and tested to record various mechanical properties such as compressive and flexural  
182 strength. Granular modelling materials were used in this investigation and Table 1 provides the  
183 material composition. The main constituents of the rock-like samples were a mixture of coarse  
184 and fine natural aggregates, approximately 5 mm and 1 mm respectively. The binder materials



185 used were ordinary Portland cement and Silica Fume. Silica Fume in fresh concrete ensures  
 186 increased cohesion and reduced bleeding. In hardened concretes, the silica fume enhances the  
 187 mechanical properties (such as compressive strength and modulus of elasticity) and reduces  
 188 permeability. Polystyrene cubes of 0.001 m<sup>3</sup> by volume were used as the mould for sample  
 189 preparation.

190 *Table 1. Composition of workpiece materials*

<b>Material</b>	<b>Low strength (LS) workpiece (kg/m<sup>3</sup>)</b>	<b>High strength (HS) workpiece (kg/m<sup>3</sup>)</b>
Portland cement (BS 12)	240	980
Silica fume	18	100
Coarse aggregate	850	850
Fine sand	275	275
Water (water/cement ratio of 0.35)	84	343

191  
 192 The cement, silica fume, coarse and fine sand were weighed out and added into a concrete  
 193 mixer rotating at low speed. Water was measured according to the cement content and added  
 194 steadily into the mixture. The mixing was continued until a desired texture was obtained. The  
 195 inside surfaces of the polystyrene cubes were coated with a thin film of mould oil to facilitate  
 196 easy removal of the mould. The mixture was filled into the mould and compacted using a steel  
 197 tamping rod in layers of 20 mm. Excess concrete was removed, and the top surface was levelled  
 198 and smoothed carefully. These moulds were left to dry at room temperature for 24 hours.  
 199 They were then de-moulded and submerged in a curing tank for a further 3 weeks. These  
 200 samples were used for compressive strength tests and for the linear cutting tests. For the  
 201 purpose of finding flexural strength and fracture toughness, rectangular moulds measuring 500  
 202 × 100 × 100 mm<sup>3</sup> were prepared separately in a similar manner.

## 203 *2.2 Specimen characterisation*

204 Compressive and flexural strength tests were conducted on the samples in order to accurately  
 205 define their mechanical properties. Compressive strength tests were conducted according to BS  
 206 EN 12390-3:2009 using the cube test specimens, while the flexural strength was conducted  
 207 according to BS EN 12390-5:2009 using the rectangular test specimen on a three-point bend  
 208 test machine (Denison Mayes Universal Testing Machine) as shown in Figure 2. A constant rate  
 209 of force was applied to the platens (approximately 3 kN/s) during the compression tests until the

210 specimen fails. The maximum load was recorded, and the compressive stress was calculated using  
211 Equation 3.

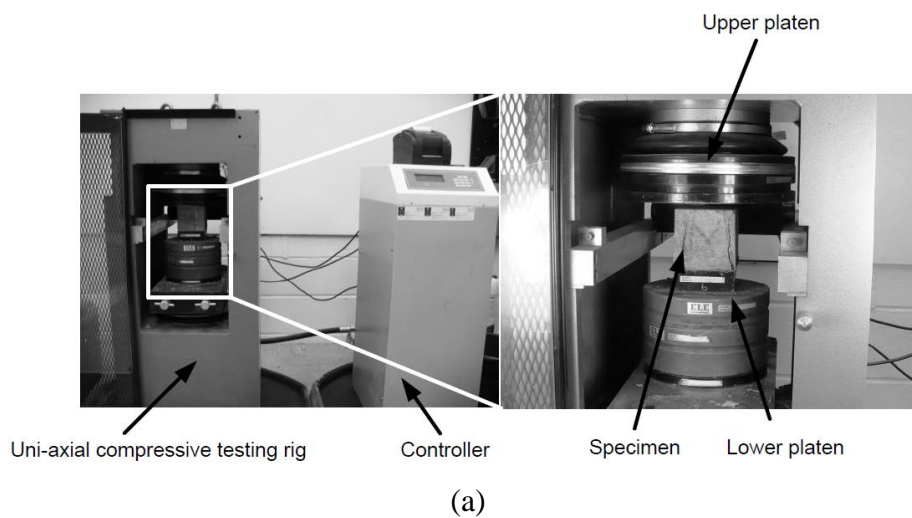
$$\sigma_c = \frac{F_c}{A_c} \quad (3)$$

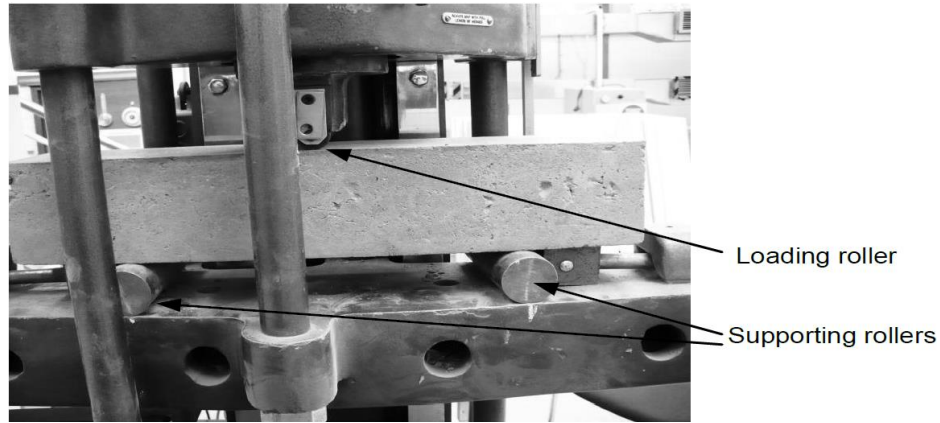
212 where  $\sigma_c$  is the compressive strength in MPa,  $F_c$  is the maximum load at failure, and  $A_c$  is the cross-  
213 sectional area of the specimen on which the force was applied.

214 During the flexural tests, the distance between the supporting rollers were maintained at 300 mm  
215 and length of the specimen was 500mm. A constant loading rate was maintained until the specimen  
216 failed, the maximum load was noted, and the flexural strength was calculated using the standard  
217 formula.

$$\sigma_f = \frac{F_f L}{2d_1 d_2^2} \quad (4)$$

218 where  $\sigma_f$  is the flexural strength in MPa,  $F_f$  is the maximum load in N,  $L$  is the distance between  
219 supporting rollers in mm, and  $d_1$  and  $d_2$  are the width and height of the of the specimen in mm. The  
220 average results from the strength tests are shown in Table 2.





(b)

221 *Figure 2. Experimental set-ups for (a) compressive strength test and (b) flexural strength test*

222

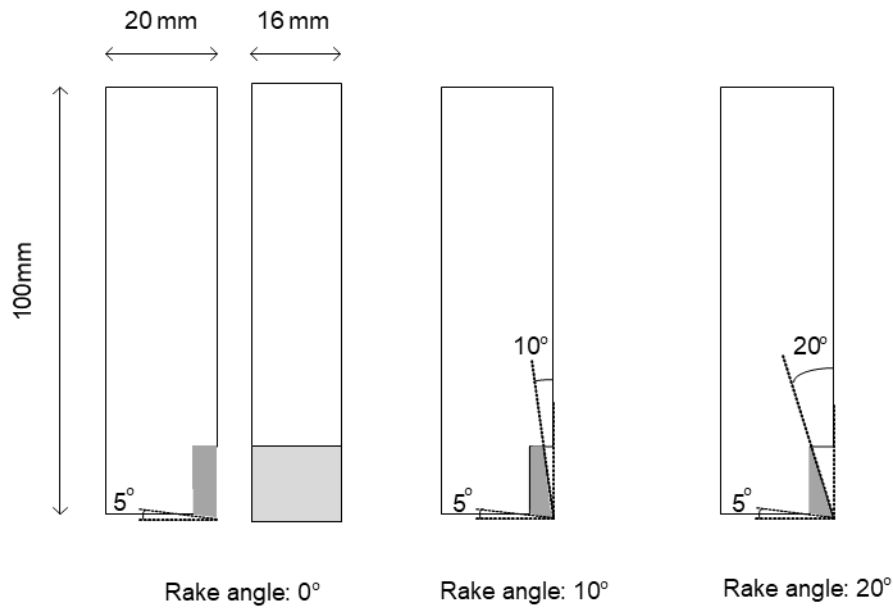
223 *Table 2. Results obtained from the uni-axial mechanical strength test*

Sample type	Flexural strength		Compressive strength		Density $\rho$ (kg/m <sup>3</sup> )
	$\sigma_f$ (MPa)	Max. Flexural load $F_f$ (kN)	$\sigma_c$ (MPa)	Max. compressive load $F_c$ (kN)	
Low strength workpiece (Sample 1)	4.4	8.9	17.5	176	2170
High strength workpiece (Sample 2)	5.7	12.6	53.5	540.4	2190

224

### 225 **2.3. Cutting tool**

226 The drag tools used in this investigation were orthogonal cutting tools with brazed tungsten  
 227 carbide tips. Based on previous literature (Jonak and Gajewski, 2008; Menezes, 2016; Richard  
 228 et al., 2012; Tiryaki and Dikmen, 2006; Yadav et al., 2018), three cutting tools were designed  
 229 and used for the cutting tests, each with a 5° clearance angle and rake angles featuring 0°, 10°  
 230 and 20° as shown in Figure 3. The width, depth and length of the cutting tools used in this  
 231 investigation were 16 mm, 20 mm and 100 mm respectively.

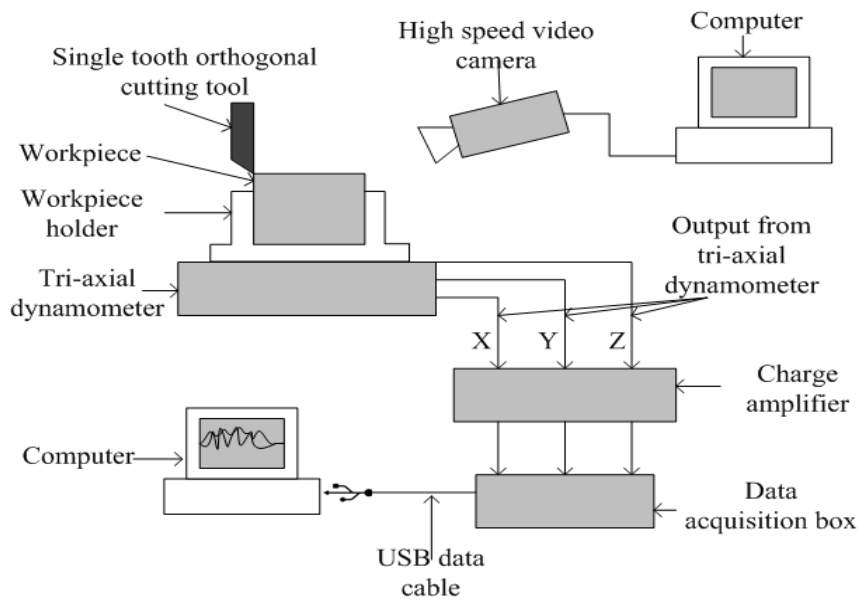


232

233 *Figure 3. Single point tungsten carbide tipped drag tools used for rock cutting experiments*  
 234 *(not to be scaled).*

235 **2.4. Linear scratch tests**

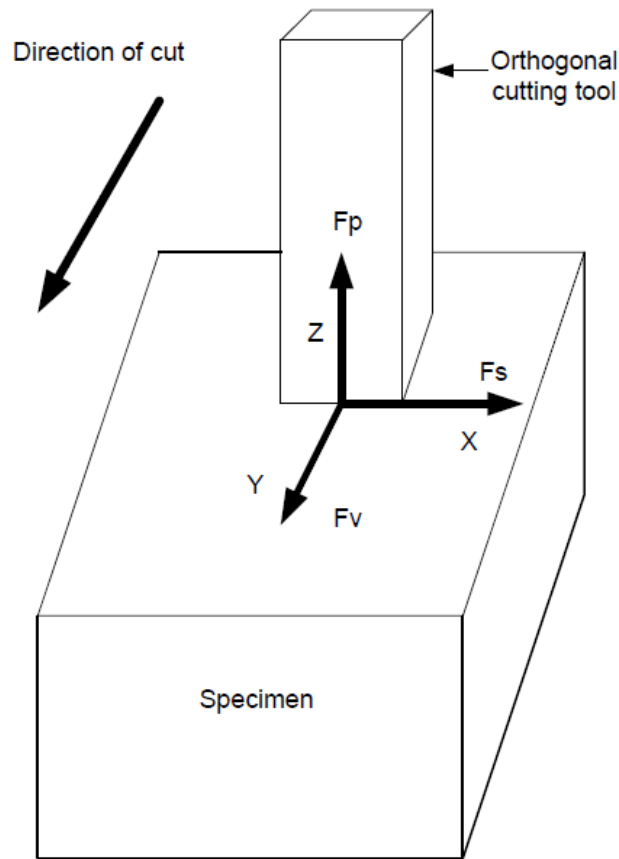
236 A test rig was designed and developed by modifying an existing shaper machine to undertake  
 237 the linear isolated (unrelieved cutting mode) scratch tests on rock-like samples using a single  
 238 point cutting tool to measure tri-axial cutting forces and to observe the fracture mechanism of  
 239 the test specimen (Aresh, 2012). The test aimed to gather data on the formation of the crushed  
 240 zone at the tip of the tool and the initiation of a crack and its propagation and the subsequent  
 241 ejection of the chip. The schematic diagram of the scratch test rig is presented and elaborated  
 242 in Figure 4. The test rig was made up of a tool holder, which held an orthogonal drag tool. The  
 243 cube shaped samples were held in place by the workpiece holder which has an adjustable clamp  
 244 to ensure a firm grip on the samples. A calibrated tri-axial dynamometer, by Kistler, was fixed  
 245 on to the table of the shaper machine, this served as a platform for the workpiece holder. The  
 246 output of the dynamometer was fed into a charge amplifier, the output of which was fed to the  
 247 computer via a data acquisition device which converted the analogue signals of the  
 248 dynamometer into digital input. High speed videos of the cutting process made it possible to  
 249 view and analyse the failure mechanism of the chip and observe the influence of the crushed  
 250 zone on the cutting process.



251

252 *Figure 4. Schematic of the test rig and force components of the tri-axial dynamometer*

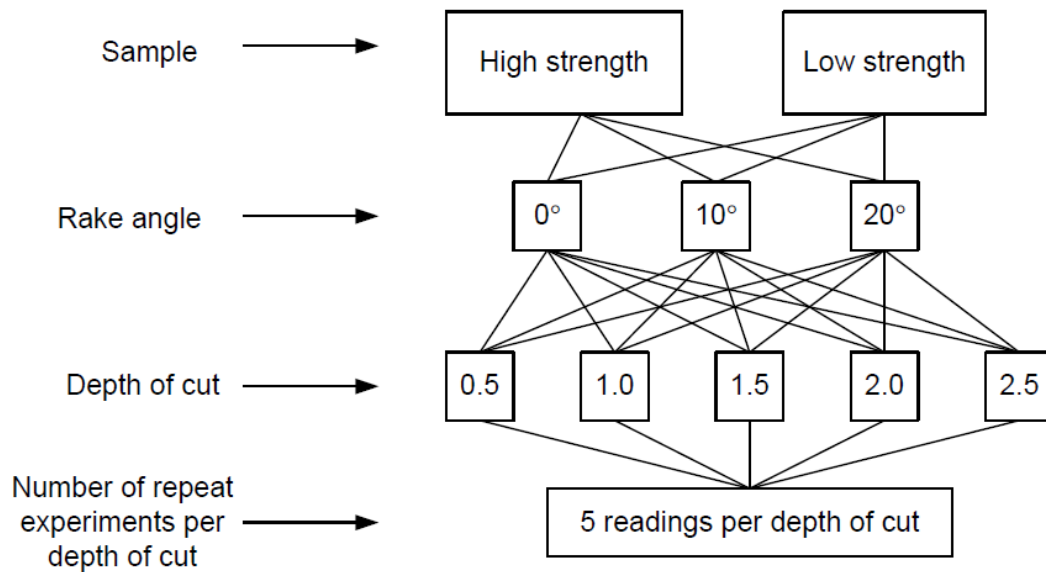
253 The tri-axial dynamometer (Kistler 9257B) measured forces along the directions of three axes,  
 254 the force components are shown in Figure 5, where  $F_p$  is the thrust force,  $F_v$  is the cutting force  
 255 and  $F_s$  is the side force. As the tool cut through the sample, the three piezo-electric transducers  
 256 in the dynamometer produce an electrical signal with a magnitude equivalent to the force  
 257 experienced at the cutting edge. This electrical signal was weak, hence it was fed into a charge  
 258 amplifier (Kistler 5010A) which amplified the signals and in turn fed them into the data  
 259 acquisition device. This data acquisition box, by National Instruments (NI USB-6221 BNC)  
 260 had 8 inputs and interfaced with a computer through a USB cable to convert the analogue input  
 261 of the charge amplifier to digital output which is read by LabVIEW software on the computer.  
 262 LabVIEW reads and logs the data; the sampling rate can be defined, and the force  
 263 measurements are logged into data files. The post-processing of this data was undertaken using  
 264 Microsoft Excel and MATLAB.



265

266 *Figure 5. Tri-axial force components measured by the dynamometer during the cutting tests*

267 Scratch tests were performed on the two sets of samples and the parameters that were changed  
 268 were the depths of cut and rake angles of the drag tool. Preliminary tests were carried out using  
 269 two cutting speeds (263 mm/s and 333 mm/s) to study the effect on the cutting force.  
 270 Subsequent scratch tests were performed using 333 mm/s velocity to limit the number of  
 271 variables in this study. The matrix for experimental tests is presented in Figure 6. The length  
 272 of cut was 100 mm. The measured cutting forces were used to calculate the SCE for a total of  
 273 150 cutting tests using Equation 1. Chips were collected and later visually analysed and  
 274 categorised according to the sample strength, depth of cut and rake angle.



275

276

Figure 6. The matrix of cutting experiments on rock-like materials

277

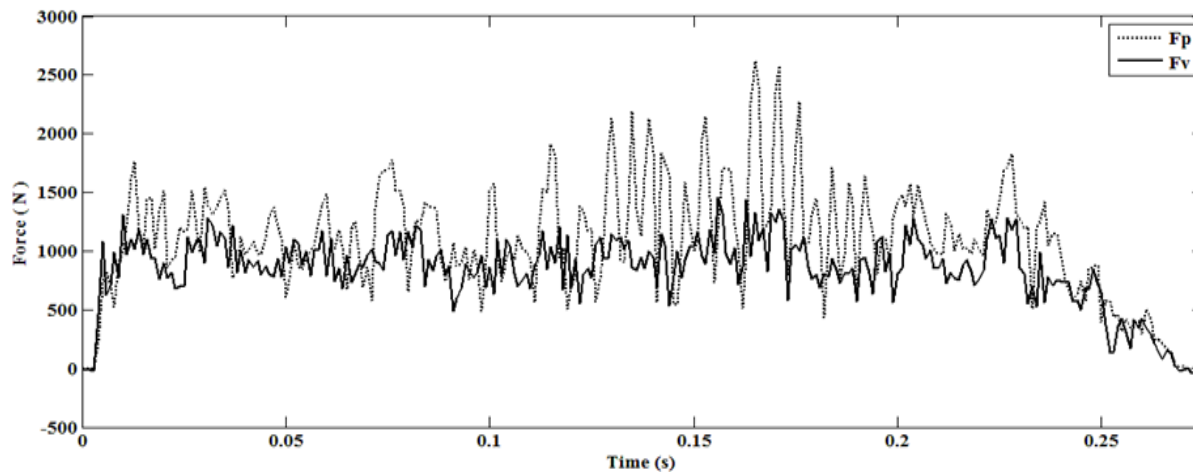
### 3. Results and Discussions

278

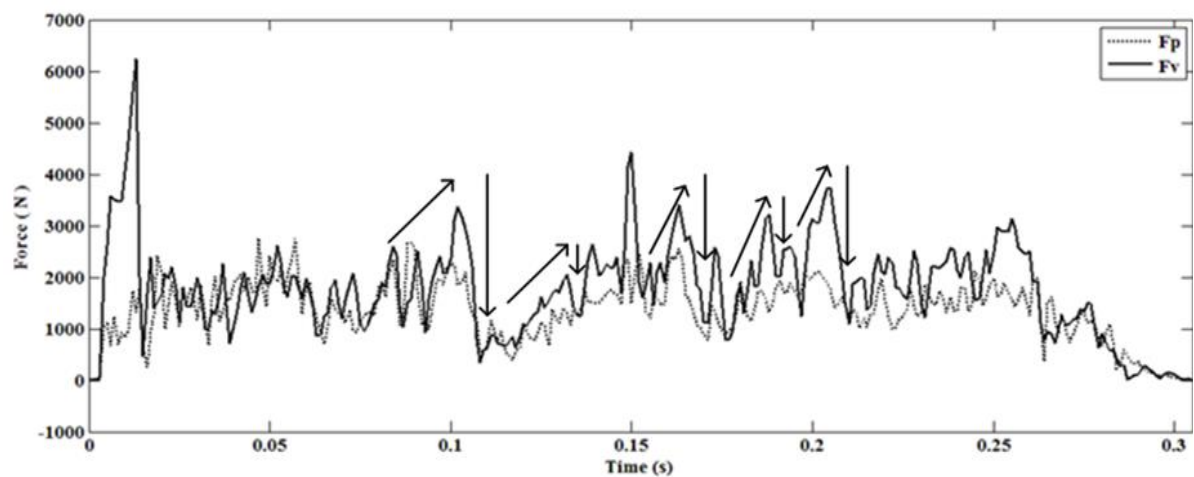
A typical force signals recorded by the dynamometer for the cutting and thrust force components at the depths of cut of 0.5 mm and 2.5 mm for the duration of the cut is presented in Figure 7. It was observed that at lower depths of cut, the cutting force ( $F_v$ ) was lower in magnitude than that of the thrust force ( $F_p$ ). However, as the depth of cut increased, this feature reversed, indicating that the cutting force turned predominant in magnitude than the thrust force. This was because at lower depth of cut the material failure mechanism was predominately ductile and hence the force required to cut was less and more akin to rubbing the surface. However, as the depth of cut increased, the failure mechanism of the material ahead of the tool tip was identified as fracture and hence the cutting force increased. This was observed for all tests irrespective of the sample type or the rake angle used in this study and also aligned with the results of another study by Mehdi *et al.* (Mohammadi et al., 2020) where cutting tests were conducted on concrete using an orthogonal cutting tool. The nature of these signals was representative of the brittle nature of the workpieces; it could be observed that as soon as the tool impacted with the rock-like workpiece, there was a gradual but a sudden rise in the cutting force (approximately 1000 N and 6000 N for LS and HS workpieces respectively). As the tool ploughs further into the workpiece, cracks initiated usually under the tool tip and propagated down before turning up and reaching the free surface thus producing a chip and the cutting force suddenly dropped. Ejection of the chip from the surface took place

295

296 usually at high speed up to 4 m/s as recorded by the use of high-speed video. This high-speed  
 297 ejection of the chip was represented by a spike in the thrust force. This cycle of local maxima  
 298 and minima of the cutting and thrust forces repeated for the entire duration of the cut, signifying  
 299 the brittle breaking off of the chip from the surface of the workpieces.



(a)



(b)

300 *Figure 7. Variation of the cutting and thrust force with increase in depths of cut (a) 0.5 mm*  
 301 *and (b) 2.5 mm*

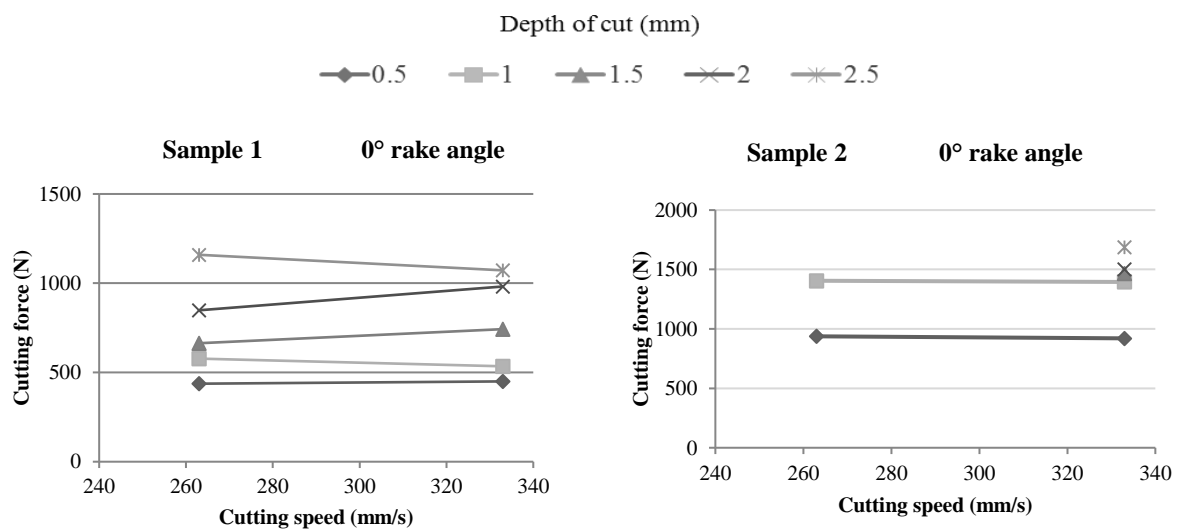
302 At shallow depths of cut (less than 1 mm) the cutting and thrust force signals were observed to  
 303 be continuous, as seen in Figure 7(a), while at greater depths, the force signals take the form  
 304 of a 'saw-tooth' profile, as seen in Figure 7(b) indicated by the arrows. The cutting events could  
 305 be distinctly recognised, for example, between 0.1s and 0.15s in Figure 7(b) the cutting force  
 306 was seen to gradually rise even though it was interspersed with local maxima and minima,  
 307 indicating the formation of minor chips and the crushed zone ahead of the tool tip. The crushed  
 308 zone was a region of highly compacted powdered material and crucial for the transmission of

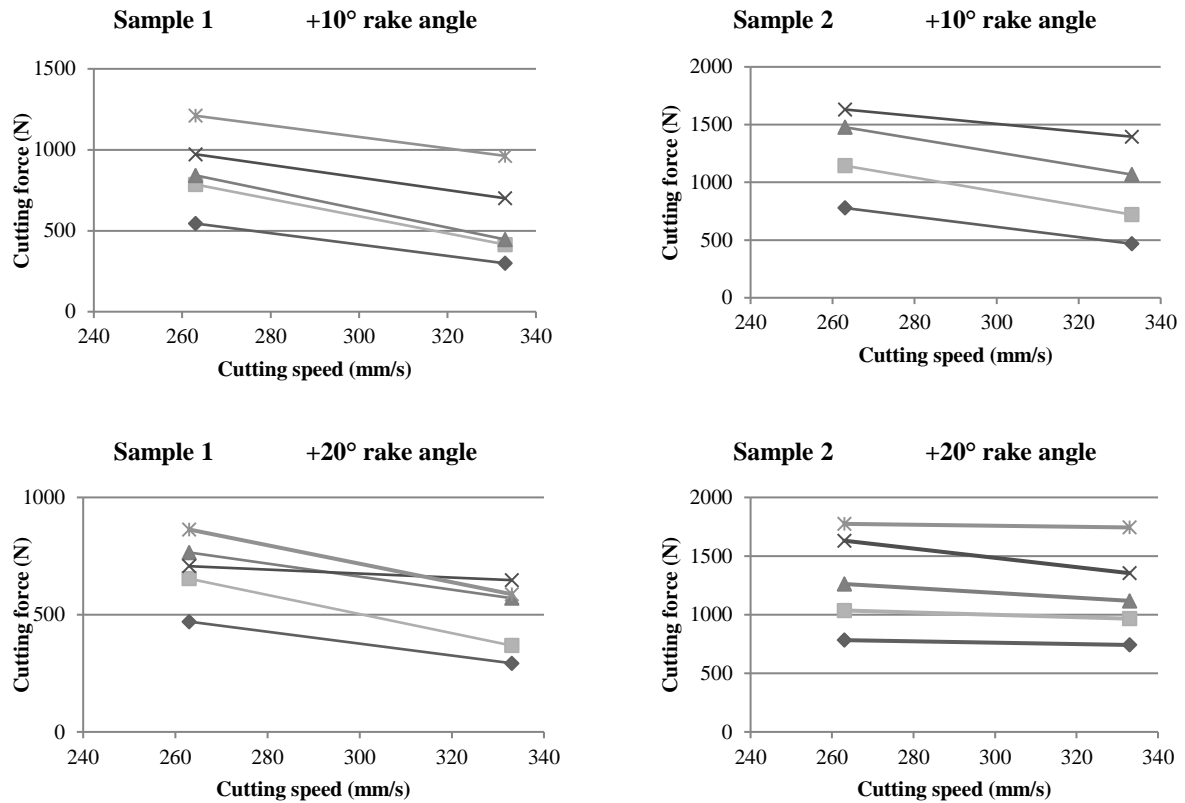


309 the cutting force from the tool to the workpiece and the saw-tooth profile of the cutting force  
 310 was characteristic of the constant build-up and breaking-off of this crushed zone as observed  
 311 using high speed video recordings.

312 **3.1. Effect of cutting velocity on the cutting force**

313 Supplementary experiments were conducted to verify the influence of the cutting speeds on the  
 314 cutting force at the depths of cut, and for the rake angles and workpiece materials used. The  
 315 shaper machine had two-speed settings of 263 mm/s and 333 mm/s. Figure 8 provides the  
 316 results of the experiment with the two speed settings. In general, at higher rake angles (10° and  
 317 20°), the cutting forces decreased with the increase of the cutting speeds for all depths of cut  
 318 and for both materials. However, at 0° rake angle, no noticeable changes were seen in the  
 319 cutting forces with an increase in the cutting speeds. Furthermore, it was observed that data  
 320 was unavailable for depths greater than 1.5 mm at the slower speed setting of 263 mm/s for the  
 321 HS workpiece specimen. This was due to the fact that the machine stalled and failed to  
 322 complete the cut. Therefore, lower cutting speed was not considered for developing a reliable  
 323 cutting model.



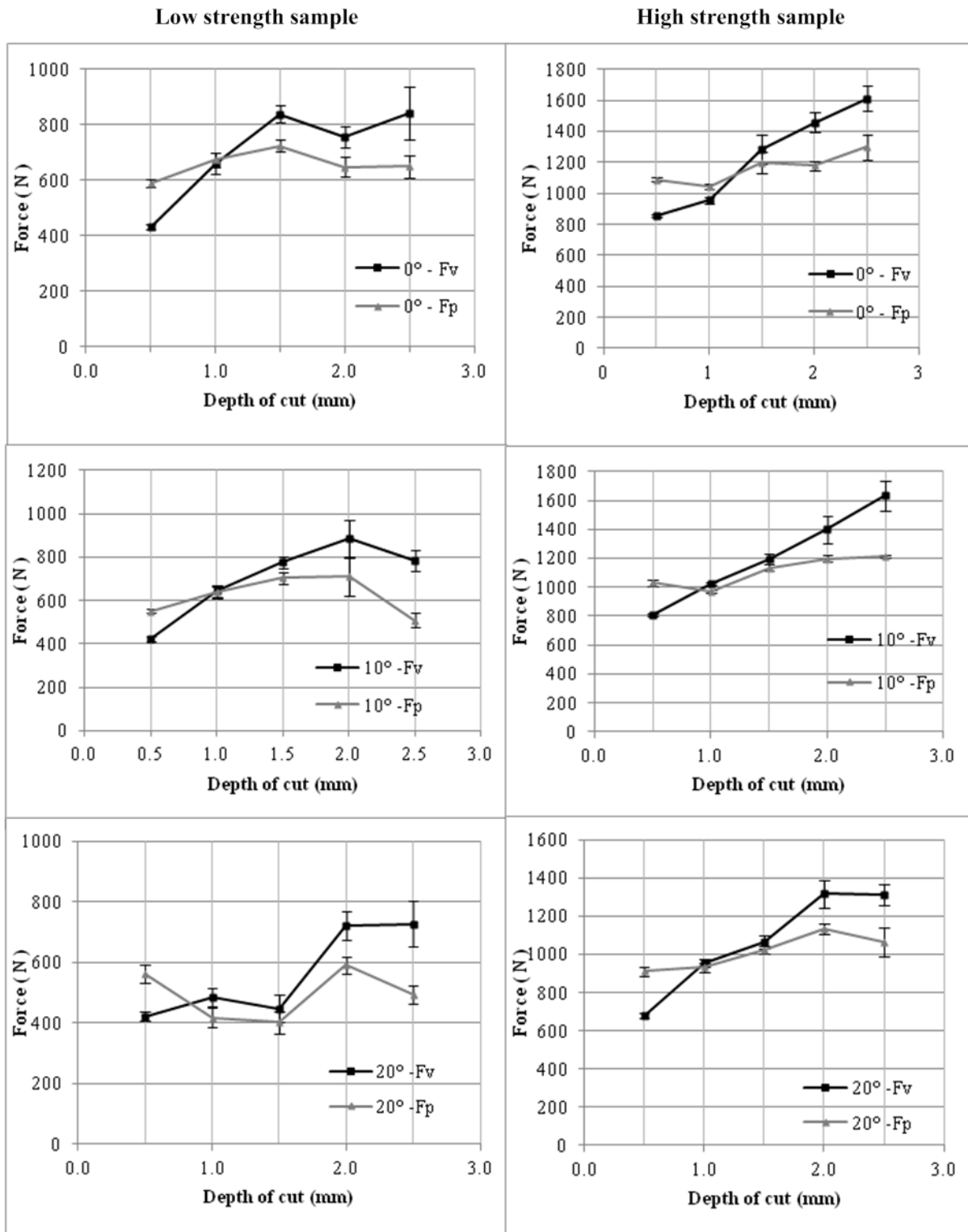


324 *Figure 8. Variation of the cutting force against change in cutting velocity*

325 **3.2. Effect of depth of cut and rake angle on the cutting force**

326 Figure 9 shows the  $F_v$  and  $F_p$  variations against the depths of cut for the two different  
 327 workpieces and three different rake angles. The cutting force increased with an increase in the  
 328 depths of cut. However, the thrust force component was seen rather to fluctuate around the  
 329 mean force of 600 N for the LS workpiece and around 1000 N for the HS workpiece and thus  
 330 remaining stable for all depths of cut irrespective of the rake angles of the cutting tools. At 0.5  
 331 mm depth of cut, it was observed that the rake angles had no significant influence. At these  
 332 shallow depths of cut the cutting action was analogous to grinding or rubbing action without  
 333 taking part in material removal action. For all rake angles, the cutting force was observed to be  
 334 lower than the thrust force. With the increase in the depth of cut, the cutting forces increased  
 335 at a higher rate than the thrust force and crossed over at a 1.0 mm depth of cut. After this cross-  
 336 over point, the differences between the cutting and thrust forces increased with an increase in  
 337 the depths of cut. This was because at lower depths of cut ( $<1$  mm), the failure mechanism of  
 338 the rock like material was in the ductile regime hence low cutting force was required but as the  
 339 depth of cut increased, the failure mechanism moved in the brittle regime hence requiring more

340 force to make the cut as evidenced by the larger debris size (Mohammadi et al., 2020; Richard  
341 et al., 2012). For the HS workpiece, the cutting and thrust forces were greatly influenced by  
342 the depth of cut particularly at the higher values. Based on the results, it could also be concluded  
343 that cutting with a 20° rake angle produced lower magnitudes of  $F_v$  and  $F_p$ , indicating that an  
344 increasing positive rake angle reduced the localised compression of the workpiece at the cutting  
345 edge, resulting in lower cutting and thrust forces (Jonak, 2001; Jonak and Gajewski, 2008;  
346 Menezes, 2016; Yadav et al., 2018).



347

348 *Figure 9. Variation of the cutting force (Fv) and thrust force (Fp) with respect to depths of*  
 349 *cut for different rake angles and workpiece materials*

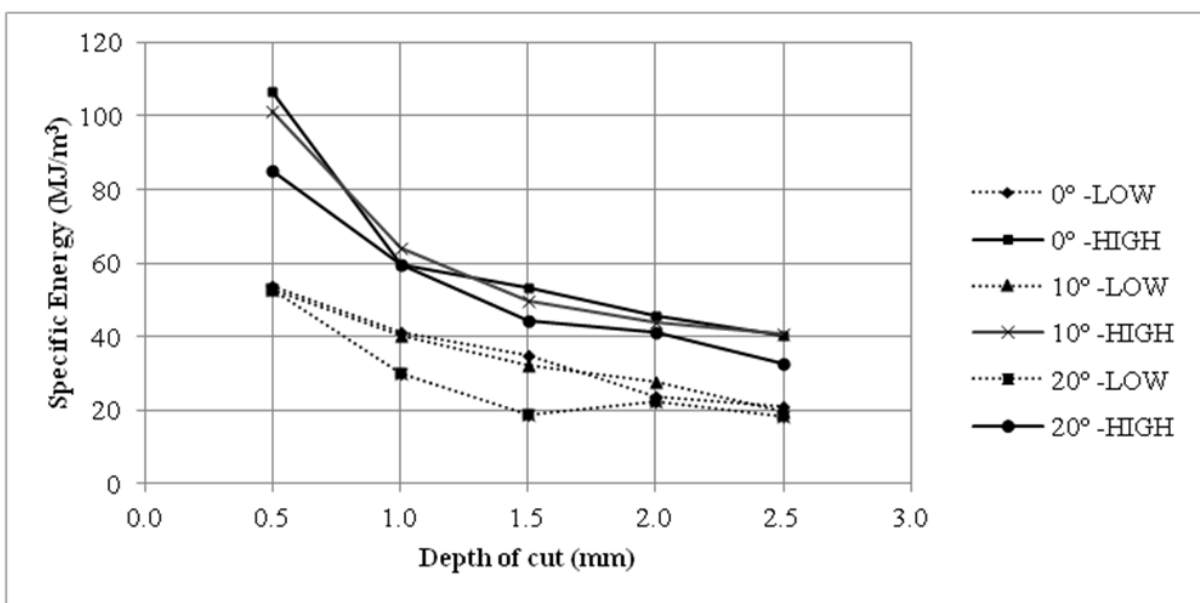
350 **3.4. Variation of specific cutting energy**

351 Figure 10 shows effect of the depths of cut on the specific energy for different rake angles and  
352 workpieces. General observations showed that SCE decreased with increasing the depth of cut  
353 irrespective of the rake angles of the cutting tool or the workpiece materials. This evidence  
354 corroborates well with other studies (Cheng et al., 2018; Mohammadi et al., 2020; Zhou et al.,  
355 2017). However, the magnitudes of the SCE for HS workpiece were higher than that of the LS  
356 workpiece. For example, SCE for the LS workpiece at 0.5 mm depth of cut was approximately  
357  $53 \text{ MJ/m}^3$  in contrast to around  $101 \text{ MJ/m}^3$  for the HS workpiece. This could be due to the fact  
358 that at shallow depth of cut, the failure mechanism was dominated by the ductile failure and  
359 the cutting force was influenced by the compressive strength of the material. At larger depths  
360 of cut, the failure mechanism was fracture and influenced by the flexural strength of the  
361 material (Richard et al., 2012). Furthermore, higher rake angle slightly reduced the SCE at  
362 various depths of cut compared to the lower rake angle. Therefore, it can be concluded that the  
363 SCE was directly influenced mainly by the depth of cut, compressive strength and flexural  
364 strength as corroborated by the other studies (Munoz et al., 2016; Richard et al., 2012).

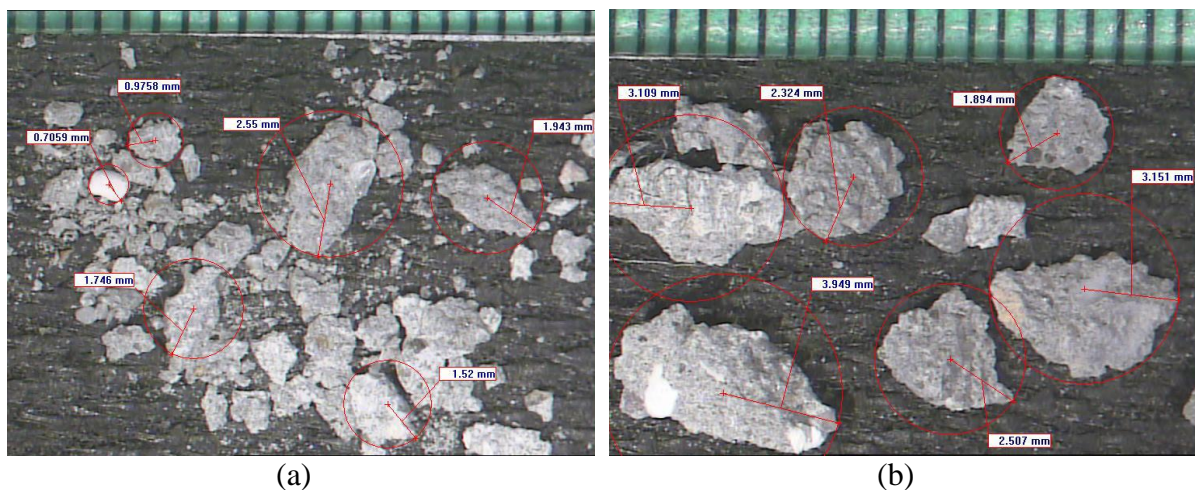
365 It should be noted that even though in this study the workpieces are classified as low-strength  
366 and high-strength, these samples actually are of very low strength (17.5 MPa) and almost  
367 medium strength (53.5 MPa) according to Deere and Miller (1966) classification. However,  
368 specific energy values calculated in this study are too high for the above soft rock-like  
369 materials. Considering quite low cutting forces generated during the tests, unexpectedly high  
370 specific energy values in this study may be attributed to the fracture properties of the artificial  
371 rock materials used. Such high specific energy values for such low unconfined compressive  
372 strength values may be obtained from evaporitic rocks like polyhalite, potash, trona, and salt.  
373 However, evaporitic rocks dominantly exhibit ductile behaviour and fragmentation is not as  
374 good as that observed in this study. Therefore, rock-like samples actually did not behave  
375 completely like a real rock material in this study.

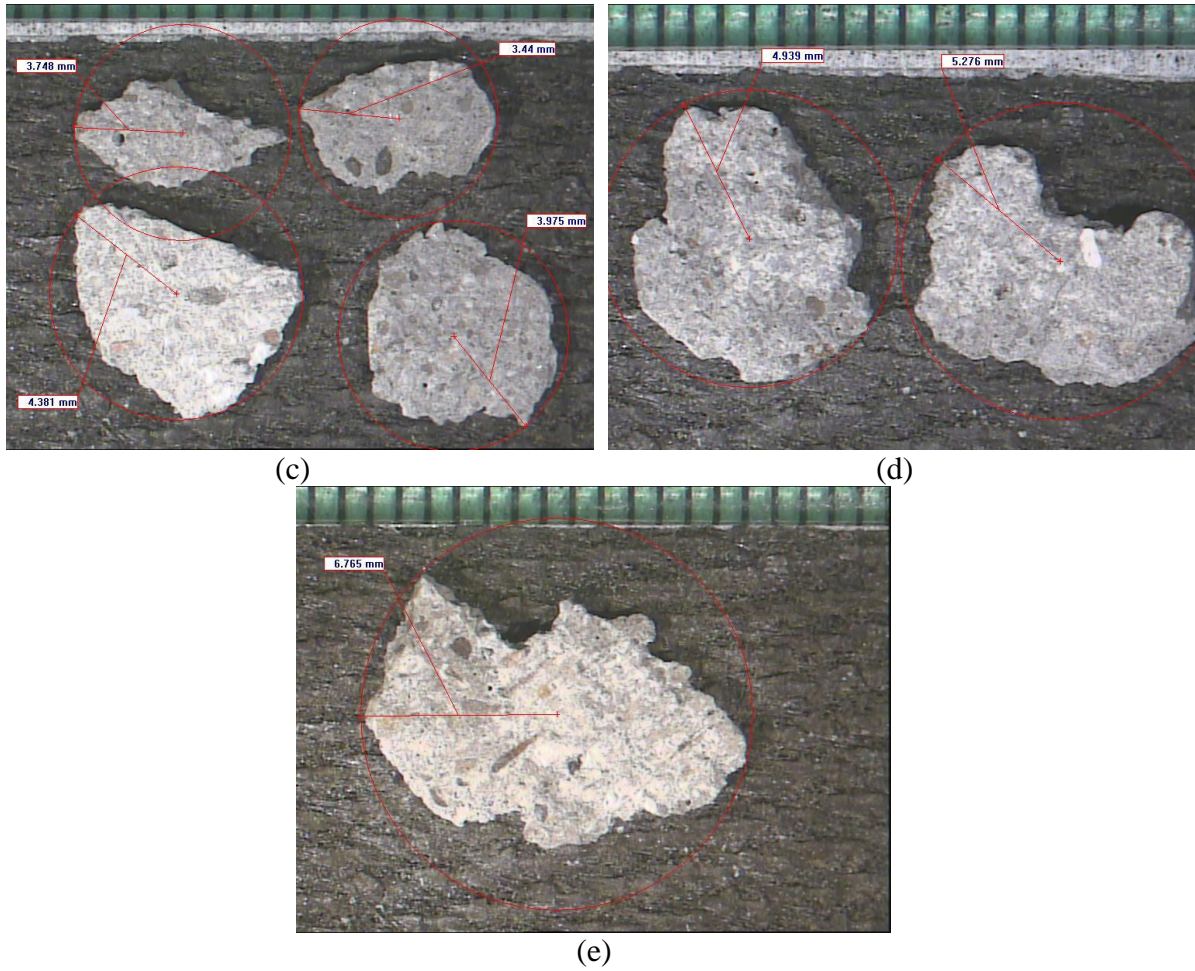
376 Cutting at a depth of 0.5 mm requires more energy as work was performed to crush and break  
377 the workpiece into fine particles rather than well-formed chips. Figure 11 shows the  
378 representative debris collected at various depths of cut using a  $10^\circ$  rake angle cutting tool on  
379 the HS workpiece. At this depth of cut, the debris was made up of fine powder and irregular  
380 fragments with a maximum diametric size measuring approximately 5 mm. As the depth of cut  
381 increased, the fine powder was observed in all the cases but the fragment size increased,  
382 reaching nearly 14 mm in diameter at 2.5 mm depth of cut, this has been observed in similar

383 rock cutting experiments performed by Cheng et al. (2018) granite sandstone and marble.  
 384 Figure 12 presents the average maximum size of the chips that were formed at various depths  
 385 of cut and rake angles for LS and HS workpieces. The SCE was observed to decrease as the  
 386 chip sizes increase, this has been observed in previous research by Mohammadi et al. (2020)  
 387 on concrete and Friant (1997) where inverse correlation was observed between chip size and  
 388 SCE. The collected chip sizes from the HS workpiece were slightly larger than those produced  
 389 from the LS workpiece for all depths of cut and rake angles. This could be attributed to the  
 390 sample strength as the LS workpiece was easily crushed than the HS workpiece despite both  
 391 being brittle in nature.

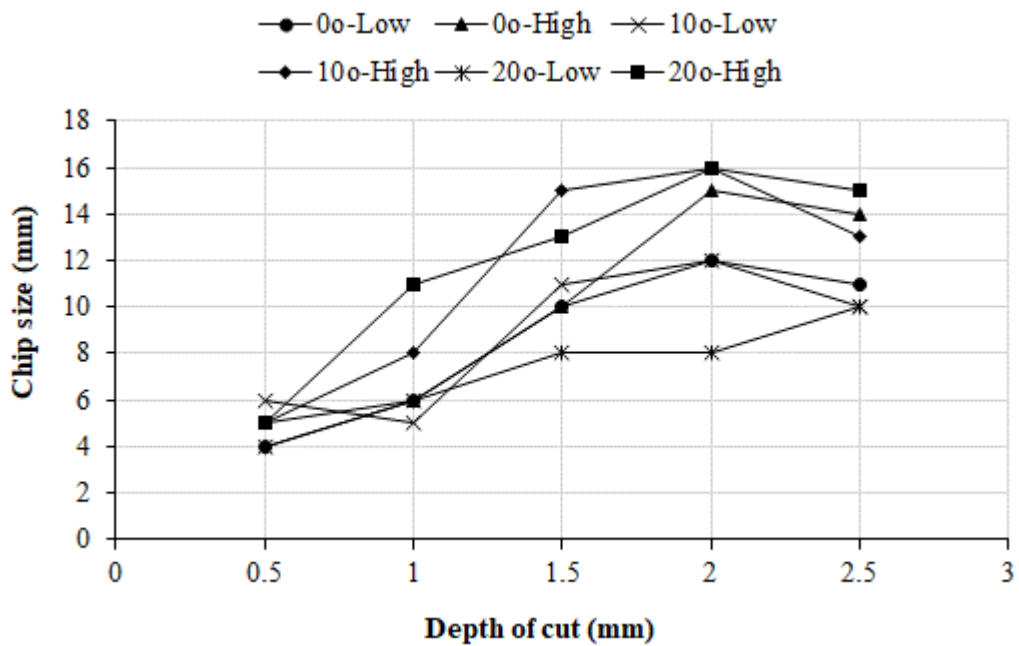


392  
 393 *Figure 10. Variation of Specific Cutting Energies with depth of cut, rake angle and sample*  
 394 *workpieces.*





395 *Figure 11. Variation of chip size with respect to the depth of cut for 10° rake angle and HS*  
 396 *workpiece*



397

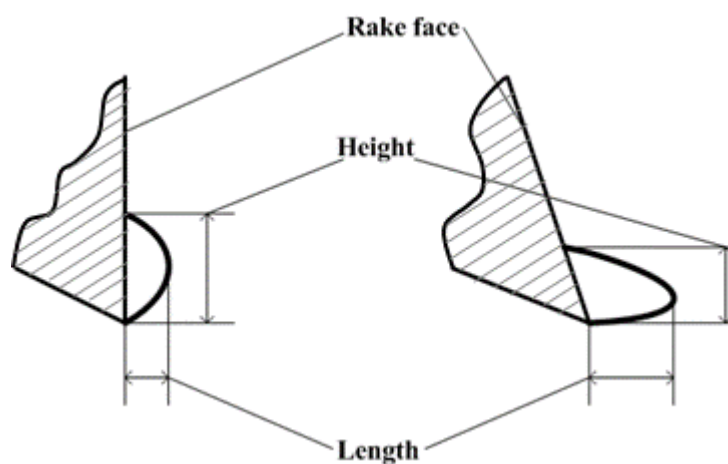
398 *Figure 12. Variation of chip size with depth of cut, rake angle and sample workpieces.*

### 399 **3.5. Effect of the Crushed Zone on the Cutting Performance of the Tool**

400 The crushed zone in rock cutting is analogous to the built-up edge in metal cutting, it is crucial  
401 to the force transmission from the tool to the rock by building up the stress ahead of the rock  
402 to a critical point whereby cracks initiate and propagate.

#### 403 3.5.1. Characteristics of the crushed zone

404 High speed video recording of the cutting process was undertaken using a Phantom v7.3 camera  
405 produced by Vision Research. The videos were analysed using their proprietary software;  
406 features of the crushed zone were extracted and the chip removal process was carefully studied.  
407 The shape of the crushed zone was observed to be continuously evolved for the duration of a  
408 cut. First, fine crushed material adhered to the tool tip and the deposit built up until a crack  
409 forms and produced the chip, and when the chip was ejected from the surface then this crushed  
410 zone was also removed completely or partially. In order to study the geometric profile of this  
411 zone, the heights and lengths of the crushed zones were measured at the instant of initiation of  
412 the chip forming crack from the high-speed video as shown in Figure 13. High speed videos  
413 were analysed frame-by-frame and visible major crack system which led to chip formation was  
414 chosen. At the point where the crack just began to form, the dimensions of the crushed zone  
415 was obtained using the video analysis tool provided by Vision Research.

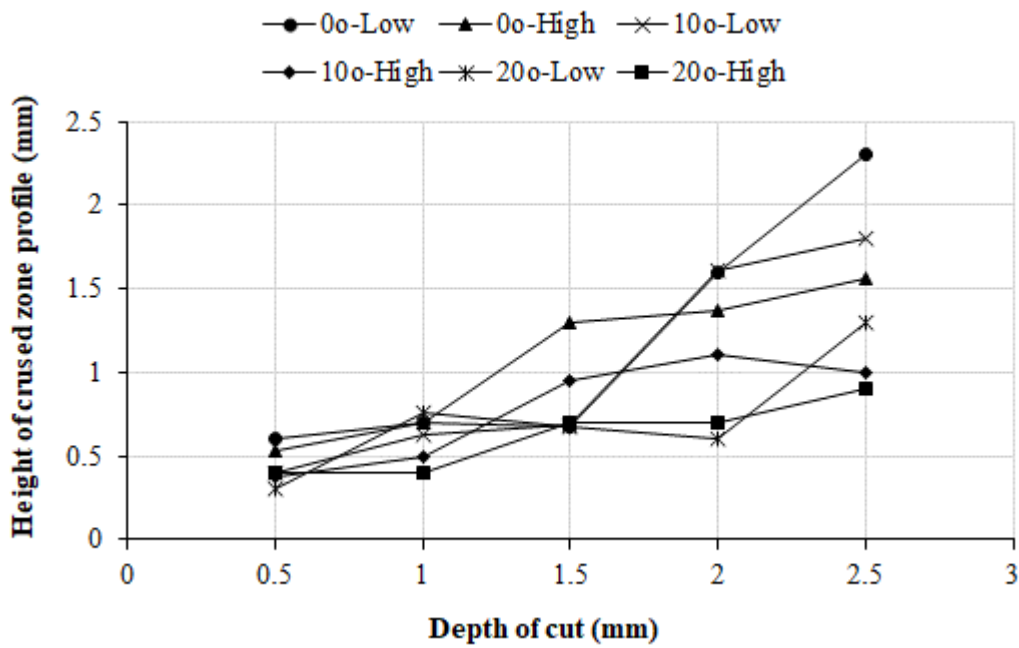


416

417 *Figure 13. Geometric profile of the crushed zone*

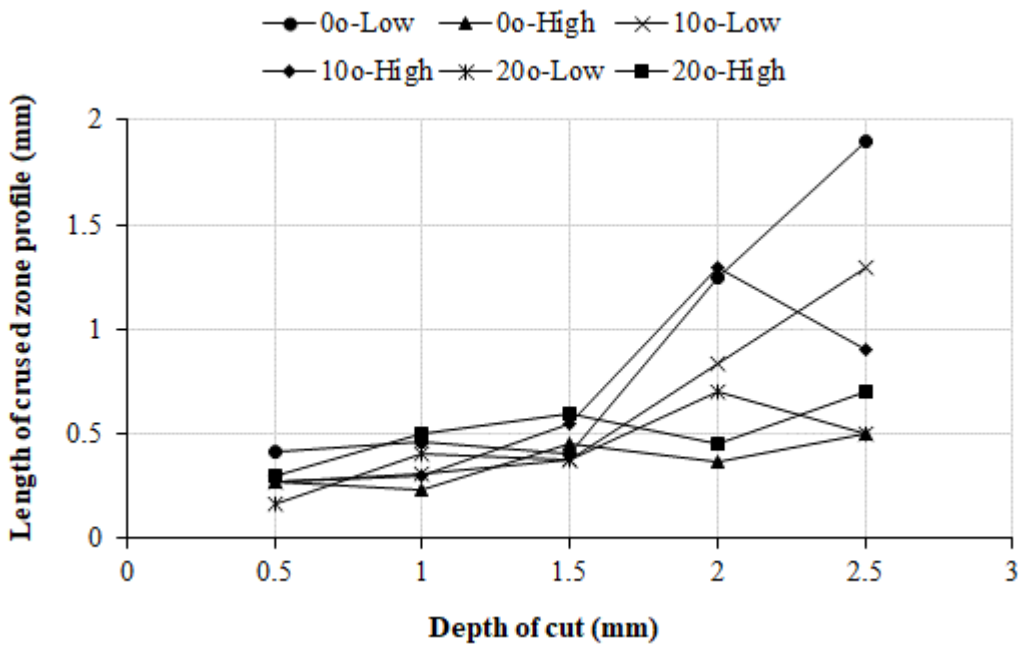


418 Figure 14 and Figure 15 show the height and length distribution of the crushed zones at  
 419 different depths of cut and rake angles for LS and HS workpieces respectively.



420

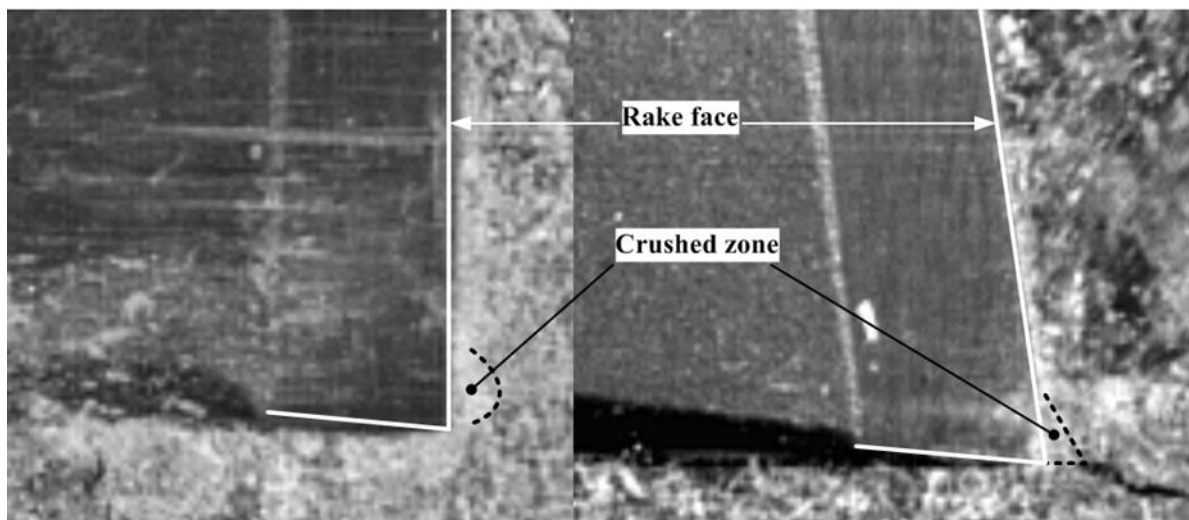
421 *Figure 14. Variation in height of crashed zone with depth of cut, rake angle and sample*  
 422 *workpieces.*



423

424 *Figure 15. Variation in length of crashed zone with depth of cut, rake angle and sample*  
 425 *workpieces.*

426 From the crushed zone measurement, it was found that with an increase in the depth of cut  
427 increased the length and width of the crushed zone. The shape of the crushed zone differed  
428 based on the rake angle of the cutting tool, as seen in other study (Yadav et al., 2018). In this  
429 study, it was observed that the  $0^\circ$  rake angle tool produced crushed zones in the shape of a  
430 hemisphere while the  $10^\circ$  and  $20^\circ$  produced wedge-shaped crushed zones as shown in Figure  
431 16. It should be acknowledged that the built-up edge was difficult to see in the still screen  
432 captures but they were clearly visible in the high-speed videos.



433

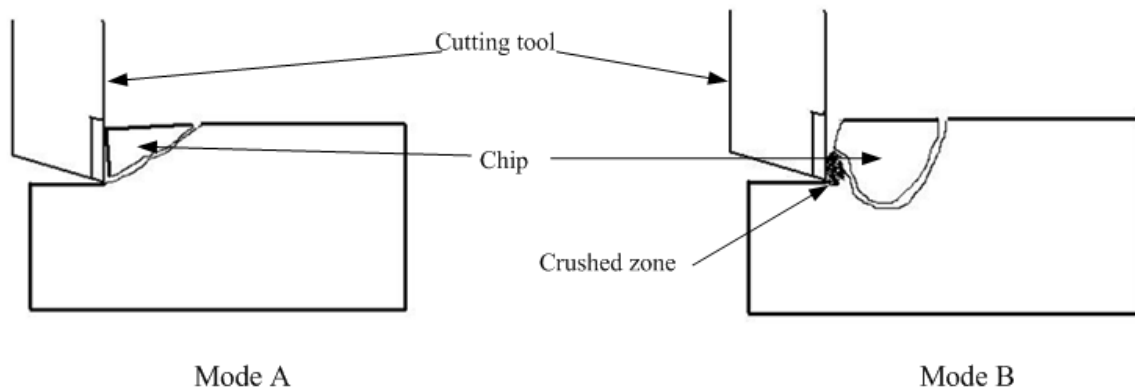
434

*Figure 16. Profile of the crushed zone for different rake angles*

435 Two kinds of chip formation process emerged, as shown in Figure 17, from the high-speed  
436 video analysis of the cutting process on both the workpieces and failure always took place due  
437 a combination of both:

438 Mode A: The chip was formed by shearing and this mode occurred in the absence of the crushed  
439 zone or when the crushed zone is just building up; and

440 Mode B: This mode of chip formation was characterised by fracture and was influenced heavily  
441 by the crushed zone.



442

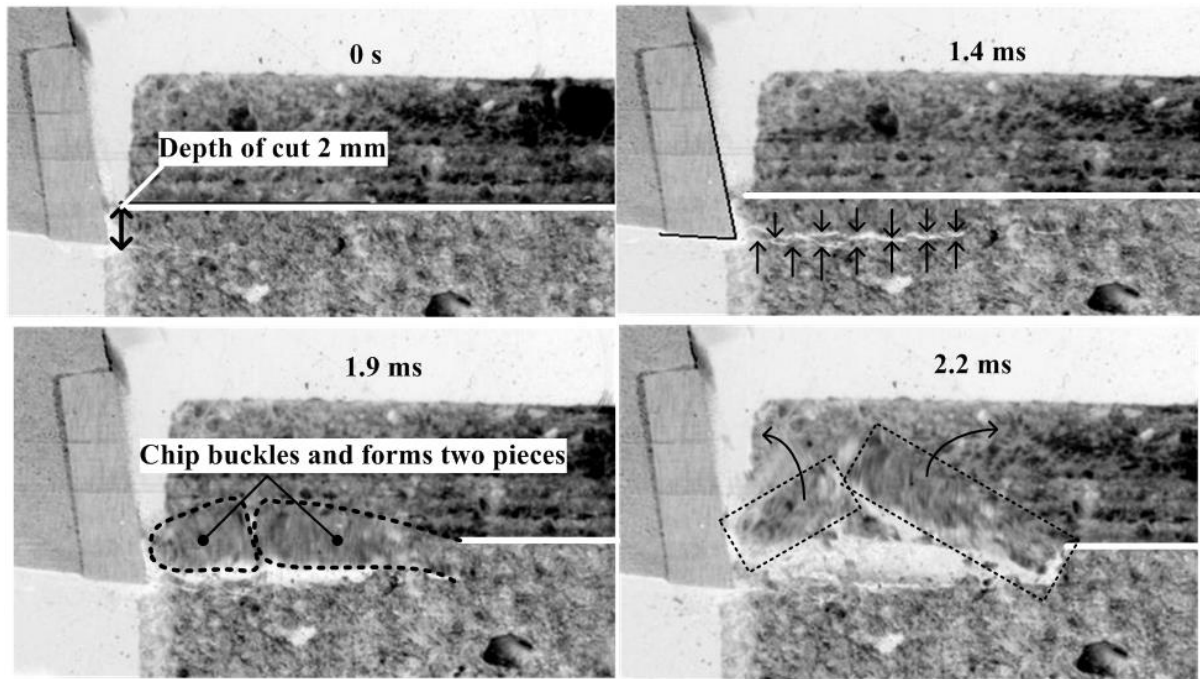
Mode A

Mode B

443

*Figure 17. The two types of chip formation mechanism*

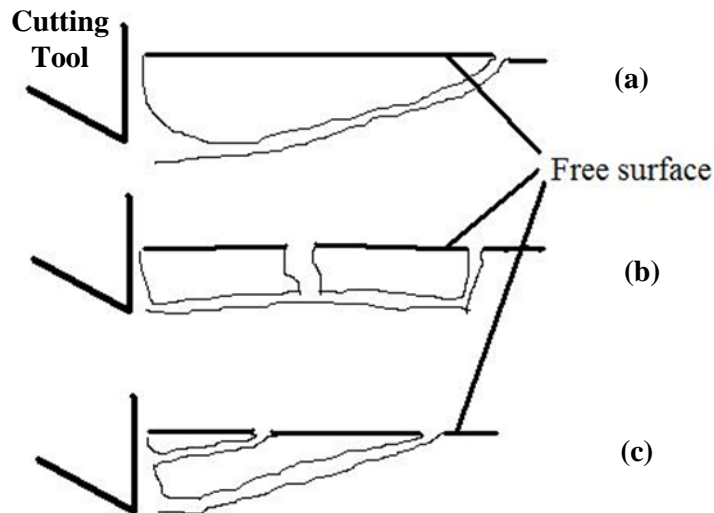
444 Deliac and Fairhurst (1988) in their rock cutting experiments involving a pick observed two  
 445 basic modes of rock failure, one is through a combination of shear/compression fracture and  
 446 the other is through fracture propagation. In this study, the direction of crack propagation in  
 447 Mode A failure were observed within a range between  $0^\circ$  -  $45^\circ$  and are not influenced by the  
 448 rake angle of the tool. When the tool made contact with the workpiece, a crack initiated  
 449 immediately ahead of the tool tip and propagated to the free surface to produce a chip. Figure  
 450 18 shows a sequence of images which shows the Mode A failure in the HS workpiece being  
 451 cut with a  $10^\circ$  rake angle cutting tool and at 2 mm depth of cut. At 0 second the tool impacted  
 452 the sample, and at 1.4 ms, cracks were found to have propagated into the sample in a direction  
 453 parallel to the cut (approximate crack length= 9 mm), as highlighted by arrows. At 1.9 ms the  
 454 separation took place and a chip was about to be formed and at 2.2 ms the cracks reached the  
 455 free surface and the chip was ejected from the workpiece surface. It was observed here that the  
 456 chip already broke into two fragments before ejection. This type of fragmentation of the chip  
 457 was observed when direction of the crack runs parallel to the free surface.



458

459 *Figure 18. Sequence of images showing the shear failure of a HS workpiece cut with a 10°*  
 460 *rake angle at 2 mm depth of cut.*

461 Different types of chip failure were observed from analysing the high-speed video and are  
 462 represented in a schematic form in Figure 19. Figure 19(a) showed the usually observed chip  
 463 with the trailing edge thicker than the leading edge. Figure 19(b) was representative of the type  
 464 of chips formed when the direction of the cracks was parallel to the free surface of the  
 465 workpiece. The chip behaved as a column and was found to buckle in the middle leading to a  
 466 splitting up of the original fragment into two. Figure 19(c) shows layered fracture usually  
 467 observed at depth of cut greater than 2 mm, usually a smaller chip layers off of the surface of  
 468 the original fragment. The latter two occurrences could be explained by the influence of  
 469 microcracks radiating away from the tool face; major cracks system coalesce with these  
 470 microcracks forming the characteristic chip as observed.



471

472

*Figure 19. Different types of chip formation observed in type A failure mechanism*

473

In Mode B failure, the chip was formed under the influence of the crushed zone. The crushed zone changed the profile of the tip of cutting tool, thus the original rake angle and the sharp cutting edge became blunt by the crushed zone. The crushed zone as discussed earlier was observed to take primarily two different shapes based on the cutting tool geometry: a hemispherical shape when cutting with a rake angle of  $0^\circ$  and a wedge shape when rake angle is greater than  $0^\circ$ . The shape and size of the crushed zone were critical to the transmission of the force from the tool to the workpiece.

480

From Figure 16 and Figure 18, it was observed that the crushed zone was formed when fine powdered samples were compacted together to form a dense clump along the tip of the tool. The fine particles flew above or below this dense region and slowly began to adhere to it thus increasing the size of the crushed zone. This crushed material created a region of intense stress and when this reached a critical limit then a crack forms on the upper level of this crushed zone and quickly propagated down into the workpiece and around the crushed zone. It usually propagated below the depth of cut leading to an overcut and then propagated to the free surface and resulted in the formation of the chip.

488

### 3.5.2. Force analysis

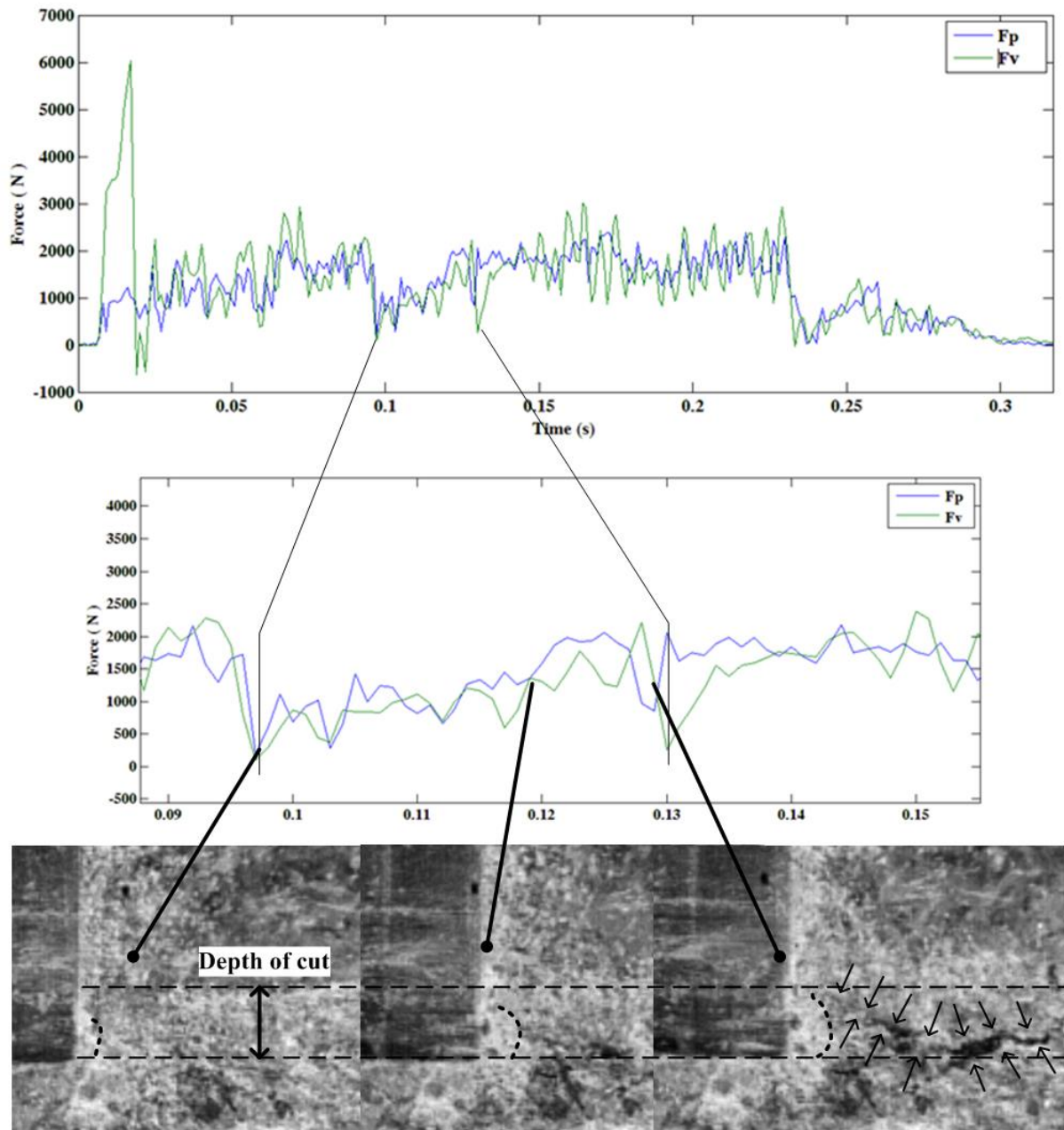
489

The crushed zone transmitted the cutting force from the tool to the workpiece. To analyse this event, force trace signals were corroborated with the high-speed video images. Figure 20 shows the cutting and thrust force traces for the HS workpiece being cut with the  $0^\circ$  rake angle tool at

490

491

492 2.5 mm depth of cut. A section of the force trace was highlighted and shown in the figure along  
493 with a sequence of images taken during that period. The force trace duration of interest was  
494 between 0.1s and 0.13s. It was seen that the cutting force gradually increased from 500 N to  
495 2000 N and so did the thrust force. The signals are interspersed by local maxima and minima  
496 peaks during this duration, this is due to microcracks opening up but not leading to a chip  
497 formation. This building up of the cutting force and the thrust force coincided with the change  
498 in size of the crushed zone as seen in Figure 20. The crushed zone was highlighted by a dashed  
499 curve which was seen to evolve in size until a critical point was reached which led to a fracture  
500 shown in the last sequence on the right (arrows highlight the path of the crack). Further  
501 observation from the video showed that the finely crushed powder ‘flow’ around the crushed  
502 zone all the while compacting it until a critical stress was reached which resulted in fracture  
503 propagation. Although Figure 14 and Figure 15 showed some loose trend between the depth of  
504 cut and the dimensions of the crushed zone but not significant enough to justify a trend.



505

506

*Figure 20. Influence of the crushed zone on the cutting force.*

507

#### **4. Numerical modelling and simulation**

508

A computational model was developed to simulate the rock cutting experiment for further

509

understanding of chip formation mechanism and corroborating the experimental observation.

510

The model used material properties that were found experimentally in this study. Material input

511

parameters were Poisson's ratio, elastic modulus and density. Table 3 lists the material

512

properties for the two types of workpieces and the cutting tool. The material property for the

513

cutting tool was selected so as to make it very stiff (Cai and Kaiser, 2004).

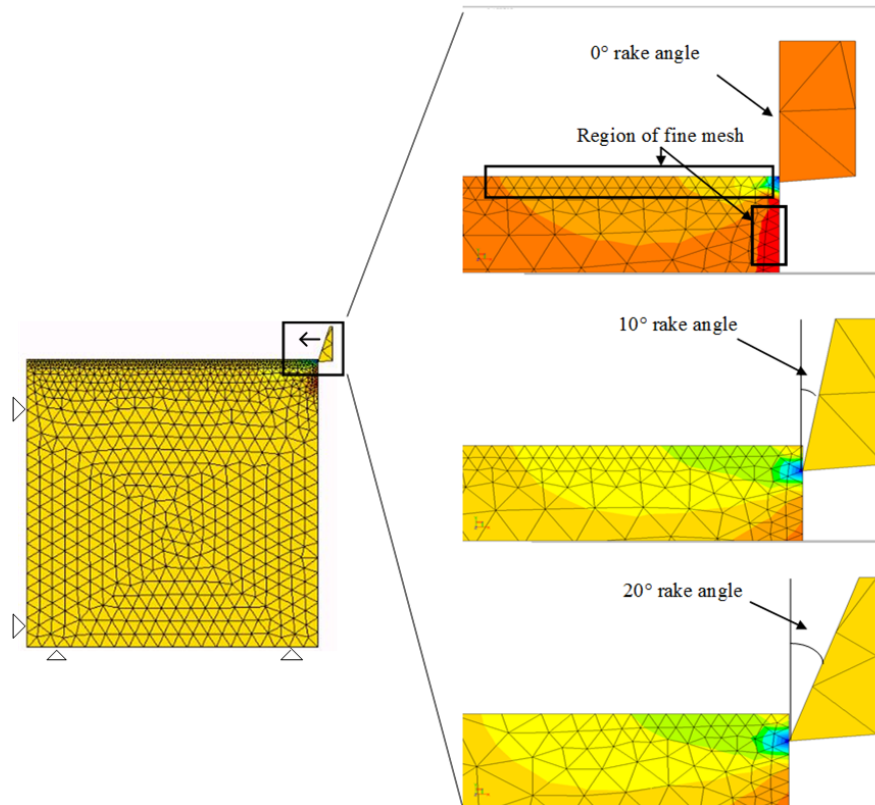
Table 3. Material properties used in numerical simulation

Material Property	Value		
	Cutting tool	HS workpiece (Sample 1)	LS workpiece (Sample 2)
Young's Modulus (GPa)	211	19.4	34.3
Poisson's Ratio	0.286	0.27	0.27
Density (kg/m <sup>3</sup> )	7838	2170	2190

515

516 ELFEN explicit solver was used in this research as it was best suited to simulate non-linear  
517 fracturing simulations. The 2D geometric model was created in ELFEN by defining points  
518 through inputting co-ordinates in the XY plane and then connecting the points by lines. The  
519 workpiece dimensions were similar to the experimental workpiece dimension, that is, 100 mm  
520 × 100 mm. The points were joined together with lines and the workpiece surface was created  
521 by the area enclosed by the lines. The cutting tool was defined in a similar manner with four  
522 points, lines and a surface. The cutting tool dimensions were changed according to the rake  
523 angles used. The boundary conditions were based on the way the workpiece was secured and  
524 the tool held in the tool holder in the experimental test rig. The workpiece had structural fixities  
525 applied to the lines on the left and the bottom, and the cutting tool had constraints applied to  
526 prevent it from moving in either direction along the y-axis and from rotating about the z-axis.  
527 Figure 21 shows the geometric model of the workpiece, cutting tool and the constraints. A  
528 mechanical load was applied on the workpiece using a velocity load assigned to the cutting  
529 tool, which was equal to the cutting speed of the tool (333 mm/s) in a direction moving from  
530 right towards left.





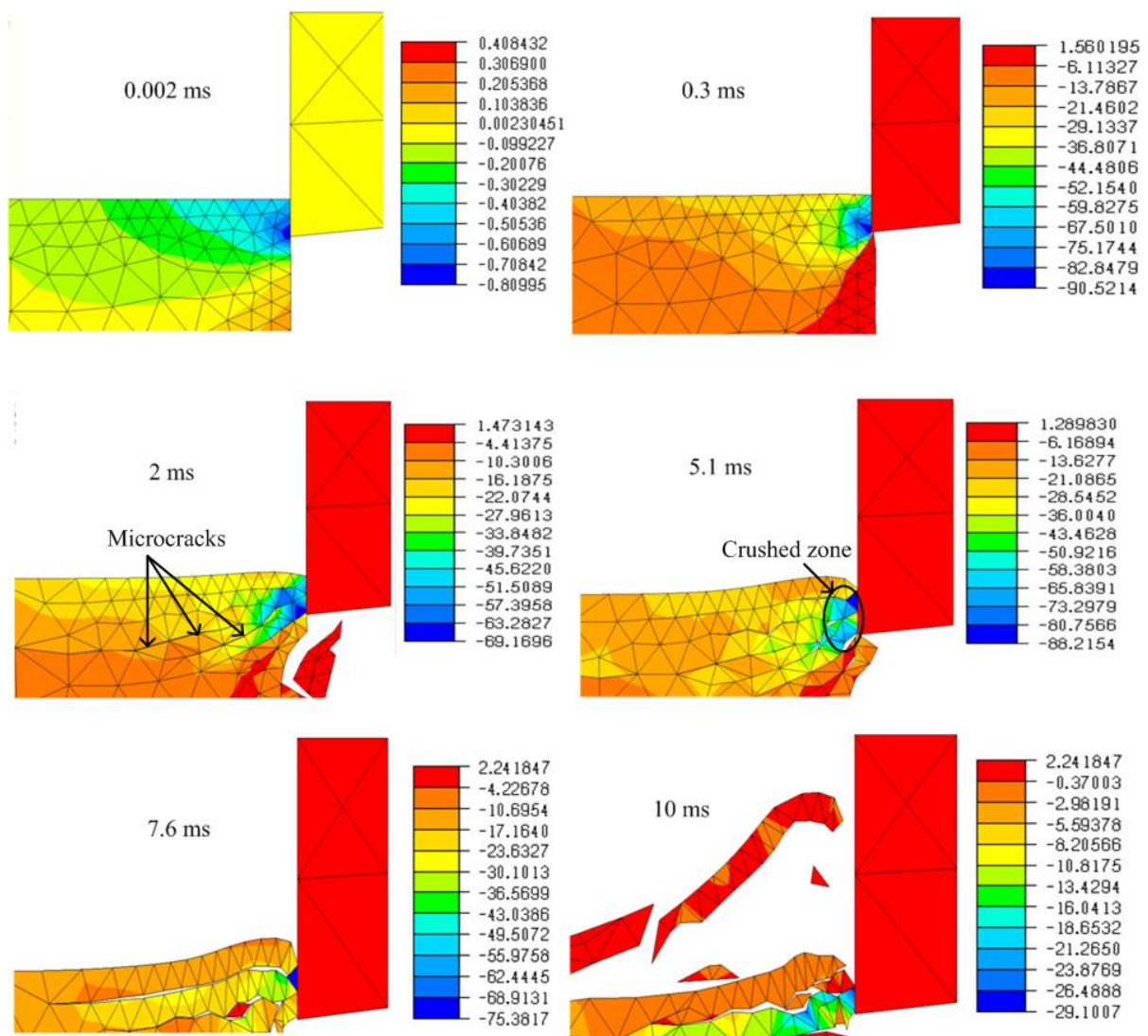
531

532 *Figure 21. Geometric model of the rock-like sample with constraints and the cutting tool with*  
 533 *different rake angles*

534 ELFEN-Explicit offers 13 types of elements covering both 2D and 3D options (Rockfield  
 535 Software Limited, 2009). In this study, a 2D linear triangular element made up of 3 nodes was  
 536 used. Each node has 2 degrees of freedom; U and V in the global coordinate system. An  
 537 unstructured mesh was generated using linear triangular elements with a side length of the  
 538 element of 5 mm for both the workpiece and the tool, but a finer mesh with side length of 1  
 539 mm was generated in the region where the tool interacts with the rock. This was done to  
 540 produce realistic fracture patterns since the fracture depends on the mesh size and density. The  
 541 work of Cai and Kaiser (2004) states that crack propagation is dependent on mesh size and  
 542 density. In a more recent work, van Wyk et al. (2014) also reflects upon the impact of changing  
 543 mesh size has on the results while Jaime et al. (2015) used the size of the rock particle to define  
 544 the mesh size. As these numerical simulations form an important tool to study the  
 545 micromechanics, it was considered important to simulate as close as the experimental  
 546 observations, hence the debris size collected from the experiments were used as basis of mesh  
 547 density, at low depth of cut, fine debris approximately 1 mm in diameter were observed, hence  
 548 the mesh size at the surface was set to 1 mm and at higher depth of cut debris size of

549 approximately 5mm were observed. The images of the simulation result given are a zoomed in  
 550 version but when the animation of the results was compared with the high-speed video  
 551 recordings of the experiments, similarities were observed in crack propagation thus justifying  
 552 the mesh choice. The cutting tool does not require a fine mesh at the contact regions, and hence,  
 553 only a few elements with side length of 5 mm were used for mesh generation.

554 Figure 22 presents a sequence of the simulation of cutting through LS workpiece. At 0.002 ms  
 555 the tool was seen to contact the workpiece resulting in a region of high compressive stress,  
 556 which further shrank and concentrated itself to the tool tip as seen at 0.3 ms. At 2 ms, tensile  
 557 parting occurred and microcracks formed and were observed to propagate down into the  
 558 sample. The formation of the crushed zone was observed at 5.1 ms which led to major crack  
 559 initiation as observed at 7.6 ms. At 10 ms, the crack reached the free surface and forms a chip  
 560 which was ejected; the crushed zone of the workpiece was still seen adhering to the tool tip.

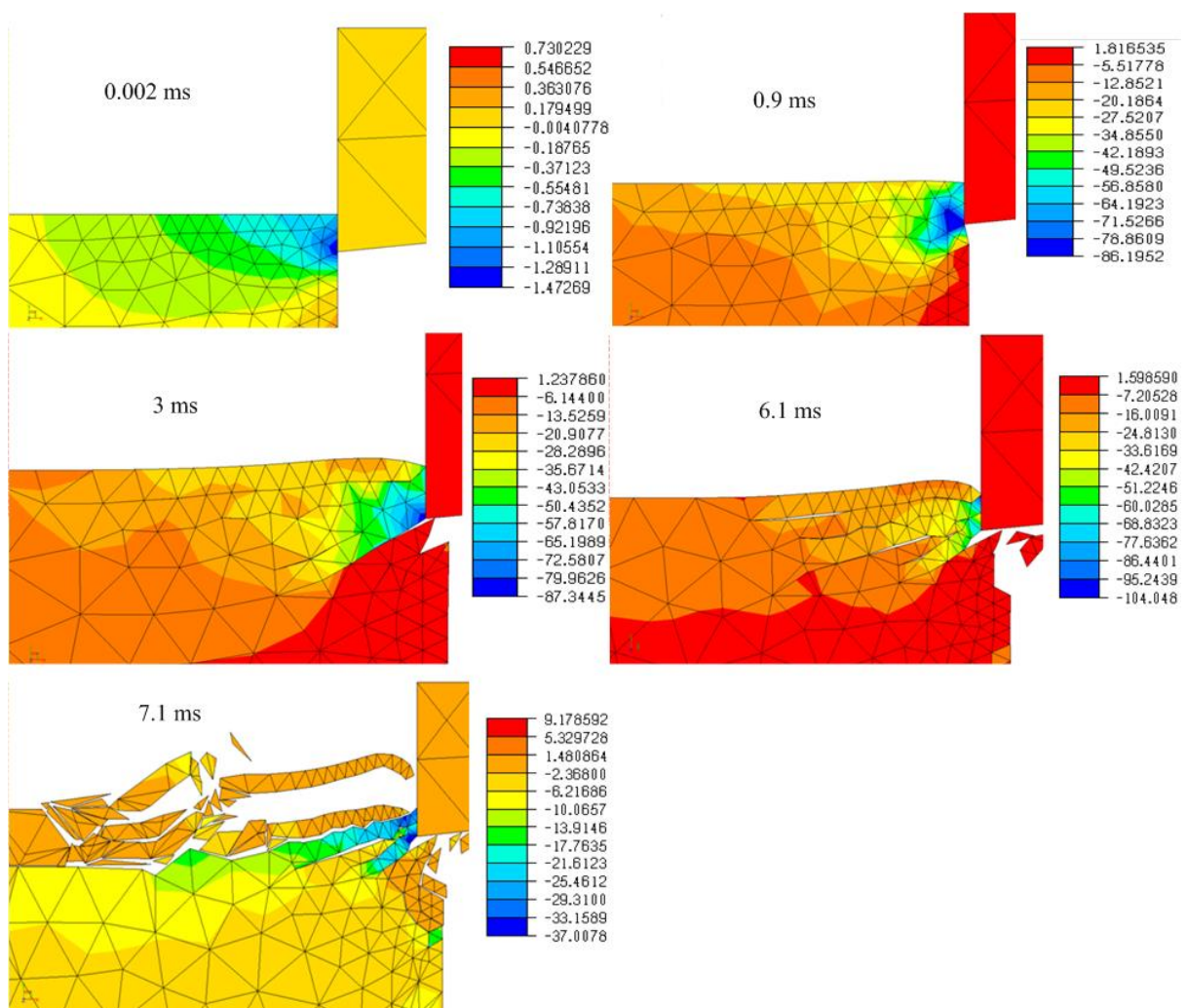


561

562 *Figure 22. Stress contours of obtained while simulating cutting through a LS workpiece at 0°*  
 563 *rake angle and 2 mm depth of cut*

564 Figure 23 shows the stress contours of the cutting simulation of the HS workpiece. It showed  
 565 cracks propagating from below the tool tip resulting in an over-cut, as seen at a time sequence  
 566 of 3 ms. At 6.1 ms, a major crack initiated from above the crushed zone and propagated into  
 567 the workpiece. At 7.1 ms, the cracks reached the free surface resulting in chip formation. The  
 568 crushed zone was seen intact.

569



570

571 *Figure 23. Stress contours obtained while simulating cutting through HS workpiece at 0°*  
 572 *rake angle and 2 mm depth of cut*

573 Numerical simulation proved to be an important tool in analysing the stress states ahead of the  
574 tool. The results of experimental and numerical simulation produced in this study were found  
575 to be similar to a previous work by Zeuch and Finger (1985) who performed cutting test using  
576 PDC cutters on three different types of rocks and came to the conclusion that the chip formation  
577 process was similar in all rock types and that fractures were nucleated at the cutter tip. They  
578 also observed an increase and a drop in the cutting force over the duration of the cut and  
579 attributed it to the formation of the crushed zone. Wei et al. (2003) performed cutting tests on  
580 Diabase and Granite and concluded that the crushed zone and the chip formation process was  
581 formed under the action of tensile stress, which was found to be common observation in all the  
582 cuts leading to the build-up of the crushed zone before the crack initiated. The state of stress in  
583 the immediate vicinity of the tool tip was critical for the formation of the crushed zone and the  
584 crack initiation. Tensile parting of the material leading to the formation of cracks and chips  
585 was found as the main mechanism in the simulations carried out in this study. High state of  
586 stress was observed during the formation of the crushed zone and microcracks were found to  
587 radiate away from this crushed zone. Crack coalescence was clearly observed in all simulations  
588 which lead to chip formation. A zone of highly pressurised workpiece at the tool tip was  
589 observed for all cuts just as the tool began to cut into the rock, immediately followed by  
590 subsurface cracks initiating above and below this zone. It was believed that the crack at the top  
591 propagated quicker than the one at the bottom, reaching the free surface and resulting in the  
592 formation of the chip. As depth of cut increased, the resulting chip size also increased, this has  
593 been observed in the works of Li et al. (2021).

## 594 **5. Conclusion**

595 In this paper, experimental tests and numerical simulations were carried out to understand the  
596 chip removal process in rock-like materials. Linear scratch tests were performed on two  
597 different workpiece samples (High Strength, HS and Low Strength, LS) using three different  
598 rake angles ( $0^\circ$ ,  $10^\circ$ ,  $20^\circ$ ) at shallow depths of cut (0.5, 1, 1.5, 2, 2.5 mm). The following  
599 conclusions can be drawn from this research:

600 1. A novel test set-up was been developed for single point cutting tests with rock like material  
601 for controlled experiments; Depth of cut and the sample strength have been identified as the  
602 major factors which influence the SCE with minor contribution from the rake angle.

- 603 2. The mean cutting force in LS workpiece was considerably less, by approximately 50% than  
604 those measured in HS workpiece for all rake angles. For both workpieces, it was observed that  
605 the 20° rake angle drag tool reduced the cutting force and thrust force needed to make a cut.
- 606 3. New chip formation mechanism is characterised by brittle failure by a combination of  
607 shearing (Mode A) and fracturing (Mode B).
- 608 4. Crushed zone geometry is influenced by the tool rake angle. The built-up edge consisted of  
609 a fine layer of crushed material changes the profile of the tool cutting edge to an apparent rake  
610 angle. This profile periodically changed due to either material deposition or detachment from  
611 the rake face.
- 612 5. Numerical simulation results supported the chip formation sequences observed during  
613 experiments vis high-speed video to a certain extent.

## 614 **References**

- 615
- 616 Acaroglu, O., Ozdemir, L., Asbury, B., 2008. A fuzzy logic model to predict specific  
617 energy requirement for TBM performance prediction. *Tunnelling and Underground*  
618 *Space Technology* 23, 600–608. <https://doi.org/10.1016/J.TUST.2007.11.003>
- 619 Aresh, B., 2012. *Fundamental Study into the Mechanics of Material Removal in Rock*  
620 *Cutting*. Newcastle upon Tyne.
- 621 Atici, U., Ersoy, A., 2009. Correlation of specific energy of cutting saws and drilling  
622 bits with rock brittleness and destruction energy. *Journal of Materials Processing*  
623 *Technology* 209, 2602–2612.  
624 <https://doi.org/10.1016/J.JMATPROTEC.2008.06.004>
- 625 O.D, A., L, D. Mechanism of rock fracture in rotary drilling using a hard alloy bit :  
626 alimov, OD frunze polyt. Inst. SU dvornikov, LT frunze polyt. Inst. SU 2F, 16R  
627 soviet min. Sci. V8, N6, 1972, P677–682 *Int. J. Rock Mech. Min. Sci. Geomech.*  
628 *Abstr.* 11, 1974, A41. [https://doi.org/10.1016/0148-9062\(74\)92814-9](https://doi.org/10.1016/0148-9062(74)92814-9).
- 629 Basarir, H., Karpuz, C., Tutluoglu, L., 2008. Specific energy based rippability  
630 classification system for coal measure rock. *Journal of Terramechanics* 45, 51–62.  
631 <https://doi.org/10.1016/J.JTERRA.2008.07.002>
- 632 Cai, M., Kaiser, P.K., 2004. Numerical Simulation Of The Brazilian Test And The  
633 Tensile Strength Of Anisotropic Rocks And Rocks With Pre-Existing Cracks.  
634 *International Journal of Rock Mechanics and Mining Sciences* 41, 478–483.  
635 <https://doi.org/10.1016/J.IJRMMS.2004.03.086>

- 636 Carpinteri, A., Chiaia, B., Invernizzi, S., 2004. Numerical analysis of indentation  
637 fracture in quasi-brittle materials. *Engineering Fracture Mechanics* 71, 567–577.  
638 [https://doi.org/10.1016/S0013-7944\(03\)00037-7](https://doi.org/10.1016/S0013-7944(03)00037-7)
- 639 Cheng, Z., Sheng, M., Li, G., Huang, Z., Wu, X., Zhu, Z., Yang, J., 2018. Imaging the  
640 formation process of cuttings: Characteristics of cuttings and mechanical specific  
641 energy in single PDC cutter tests. *Journal of Petroleum Science and Engineering*  
642 171, 854–862. <https://doi.org/10.1016/J.PETROL.2018.07.083>
- 643 Cho, J.W., Jeon, S., Yu, S.H., Chang, S.H., 2010. Optimum spacing of TBM disc  
644 cutters: A numerical simulation using the three-dimensional dynamic fracturing  
645 method. *Tunnelling and Underground Space Technology* 25, 230–244.  
646 <https://doi.org/10.1016/J.TUST.2009.11.007>
- 647 Cho, N., Martin, C.D., Segol, D.C., 2008. Development of a shear zone in brittle rock  
648 subjected to direct shear. *International Journal of Rock Mechanics and Mining*  
649 *Sciences* 45, 1335–1346. <https://doi.org/10.1016/J.IJRMMS.2008.01.019>
- 650 Copur, H., 2010. Linear stone cutting tests with chisel tools for identification of cutting  
651 principles and predicting performance of chain saw machines. *International Journal*  
652 *of Rock Mechanics and Mining Sciences* 47, 104–120.  
653 <https://doi.org/10.1016/J.IJRMMS.2009.09.006>
- 654 Deere, D.U., Miller, R.P., 1966. *Engineering Classification And Index Properties For*  
655 *Intact Rock*. Illinois.
- 656 Deliac, E.P., Fairhurst, C.E., 1988. Theoretical and practical investigations of improved  
657 hard rock cutting systems, in: *The 29th U.S. Symposium on Rock Mechanics*  
658 *(USRMS)*.
- 659 Ersoy, A., Waller, M.D., 1995. Wear characteristics of PDC pin and hybrid core bits in  
660 rock drilling. *Wear* 188, 150–165. [https://doi.org/10.1016/0043-1648\(95\)06646-2](https://doi.org/10.1016/0043-1648(95)06646-2)
- 661 Friant, J.E., 1997. Disc cutter technology applied to drill bits, in: *Natural Gas*  
662 *Conference: Emerging Technologies for the Natural Gas Industry*.
- 663 Huang, S.L., Wang, Z.W., 1997. The mechanics of diamond core drilling of rocks.  
664 *International Journal of Rock Mechanics and Mining Sciences* 34, 134.e1-134.e14.  
665 [https://doi.org/10.1016/S1365-1609\(97\)00233-5](https://doi.org/10.1016/S1365-1609(97)00233-5)
- 666 Jaime, M.C., Zhou, Y., Lin, J.S., Gamwo, I.K., 2015. Finite element modeling of rock  
667 cutting and its fragmentation process. *International Journal of Rock Mechanics and*  
668 *Mining Sciences* 80, 137–146. <https://doi.org/10.1016/J.IJRMMS.2015.09.004>
- 669 Jing, L., 2003. A review of techniques, advances and outstanding issues in numerical  
670 modelling for rock mechanics and rock engineering. *International Journal of Rock*  
671 *Mechanics and Mining Sciences* 40, 283–353. [https://doi.org/10.1016/S1365-](https://doi.org/10.1016/S1365-1609(03)00013-3)  
672 [1609\(03\)00013-3](https://doi.org/10.1016/S1365-1609(03)00013-3)

- 673 Jing, L., Hudson, J.A., 2002. Numerical methods in rock mechanics. *International*  
674 *Journal of Rock Mechanics and Mining Sciences* 39, 409–427.  
675 [https://doi.org/10.1016/S1365-1609\(02\)00065-5](https://doi.org/10.1016/S1365-1609(02)00065-5)
- 676 Jonak, J., 2001. Influence of Friction in the Problem of Cutting Wedge Penetration into  
677 Brittle Materials. *Journal of Mining Science* 37.  
678 <https://doi.org/10.1023/A:1015123224716>
- 679 Jonak, J., Gajewski, J., 2008. Identification of ripping tool types with the use of  
680 characteristic statistical parameters of time graphs. *Tunnelling and Underground*  
681 *Space Technology* 23, 18–24. <https://doi.org/10.1016/J.TUST.2006.12.002>
- 682 Khan, F., Daadbin, A., Persson, M., Haider, J., 2014. Development of AlTiN coated  
683 carbide bandsaw for machining titanium-17 alloy. *International Journal of Surface*  
684 *Science and Engineering* 8, 11–27.
- 685 Khan, F.N., Daadbin, A., Persson, M., Haider, J., Hellbergh, H., 2012. Assessing the  
686 performance of TiAlSiN coating on bandsaw tooth when cutting Ti-17 alloy.  
687 *Proceedings of the Institution of Mechanical Engineers, Part B: Journal of*  
688 *Engineering Manufacture* 226. <https://doi.org/10.1177/0954405411431194>
- 689 Li, Y., Chen, Z., Ye, Y., Yang, Y., 2021. Combined finite-discrete element method for  
690 modeling the interaction between single PDC cutter and brittle rock. *Journal of*  
691 *Petroleum Science and Engineering* 207, 109133.  
692 <https://doi.org/10.1016/J.PETROL.2021.109133>
- 693 Lindqvist, P.-A., Hai-Hui, L., 1983. Behaviour of the crushed zone in rock indentation.  
694 *Rock Mechanics and Rock Engineering* 16. <https://doi.org/10.1007/BF01033280>
- 695 Menezes, P.L., 2016. Influence of friction and rake angle on the formation of built-up  
696 edge during the rock cutting process. *International Journal of Rock Mechanics and*  
697 *Mining Sciences* 88. <https://doi.org/10.1016/j.ijrmms.2016.07.013>
- 698 Mohammadi, M., Khademi Hamidi, J., Rostami, J., Goshtasbi, K., 2020. A Closer Look  
699 into Chip Shape/Size and Efficiency of Rock Cutting with a Simple Chisel Pick: A  
700 Laboratory Scale Investigation. *Rock Mechanics and Rock Engineering* 53.  
701 <https://doi.org/10.1007/s00603-019-01984-5>
- 702 Munoz, H., Taheri, A., Chanda, E., 2016. Rock cutting characteristics on soft-to-hard  
703 rocks under different cutter inclinations. *International Journal of Rock Mechanics*  
704 *and Mining Sciences* 87. <https://doi.org/10.1016/j.ijrmms.2016.05.014>
- 705 Nishimatsu, Y., 1972. The mechanics of rock cutting. *International Journal of Rock*  
706 *Mechanics and Mining Sciences & Geomechanics Abstracts* 9, 261–270.  
707 [https://doi.org/10.1016/0148-9062\(72\)90027-7](https://doi.org/10.1016/0148-9062(72)90027-7)

- 708 Ozbay, M.U., Dede, T., Napier J.A.L., 1996. Physical and numerical modelling of rock  
709 fracture. *The Journal of The South African Institute of Mining and Metallurgy* 317–  
710 327.
- 711 Richard, T., Dagrain, F., Poyol, E., Detournay, E., 2012. Rock strength determination  
712 from scratch tests. *Engineering Geology* 147–148, 91–100.  
713 <https://doi.org/10.1016/J.ENGGEOL.2012.07.011>
- 714 Rockfield Software Limited, 2009. *ELFEN Explicit Manual, Version 4.4.* ed. Rockfield  
715 Software Limited, Swansea.
- 716 Rostamsowlat, I., 2018. Effect of Cutting Tool Properties and Depth of Cut in Rock  
717 Cutting: An Experimental Study. *Rock Mechanics and Rock Engineering* 51.  
718 <https://doi.org/10.1007/s00603-018-1440-2>
- 719 Rostamsowlat, I., Akbari, B., Evans, B., 2018. Analysis of rock cutting process with a  
720 blunt PDC cutter under different wear flat inclination angles. *Journal of Petroleum*  
721 *Science and Engineering* 171, 771–783.  
722 <https://doi.org/10.1016/J.PETROL.2018.06.003>
- 723 Rostamsowlat, I., Evans, B., Kwon, H.J., 2022. A review of the frictional contact in rock  
724 cutting with a PDC bit. *Journal of Petroleum Science and Engineering* 208, 109665.  
725 <https://doi.org/10.1016/J.PETROL.2021.109665>
- 726 Roxborough, F.F., 1987. The role of some basic rock properties in assessing cuttability,  
727 in: *Proceedings of the Seminar on Tunnels: Wholly Engineered Structures.* The  
728 Institute of Engineers Australia and AFCC, Sydney.
- 729 Sarwar, M., 1998. Application of advanced surface engineering treatments to multi-point  
730 cutting edges. *Surface and Coatings Technology* 108–109, 612–619.  
731 [https://doi.org/10.1016/S0257-8972\(98\)00594-5](https://doi.org/10.1016/S0257-8972(98)00594-5)
- 732 Sarwar, M., Persson, M., Hellbergh, H., 2007. Wear of the cutting edge in the  
733 bandsawing operation when cutting austenitic 17-7 stainless steel. *Wear* 263, 1438–  
734 1441. <https://doi.org/10.1016/J.WEAR.2006.12.066>
- 735 Sarwar, M., Persson, M., Hellbergh, H., Haider, J., 2009. Measurement of specific  
736 cutting energy for evaluating the efficiency of bandsawing different workpiece  
737 materials. *International Journal of Machine Tools and Manufacture* 49, 958–965.  
738 <https://doi.org/10.1016/J.IJMACHTOOLS.2009.06.008>
- 739 Sengun, N., Altindag, R., 2013. Prediction of specific energy of carbonate rock in  
740 industrial stones cutting process. *Arabian Journal of Geosciences* 6.  
741 <https://doi.org/10.1007/s12517-011-0429-x>
- 742 Stimpson, B., 1970. Modelling materials for engineering rock mechanics. *International*  
743 *Journal of Rock Mechanics and Mining Sciences & Geomechanics Abstracts* 7, 77–  
744 121. [https://doi.org/10.1016/0148-9062\(70\)90029-X](https://doi.org/10.1016/0148-9062(70)90029-X)



- 745 Suwannapini, Sompop, 1975. The mechanical and cutting properties of rocks related to  
746 the application of tunnel boring machines. Thesis. [https://core.ac.uk/display/  
747 153776290](https://core.ac.uk/display/153776290). (Accessed 6 November 2021).
- 748 Tien, Y.M., Kuo, M.C., Juang, C.H., 2006. An experimental investigation of the failure  
749 mechanism of simulated transversely isotropic rocks. *International Journal of Rock  
750 Mechanics and Mining Sciences* 43, 1163–1181.  
751 <https://doi.org/10.1016/J.IJRMMS.2006.03.011>
- 752 Tiryaki, B., Dikmen, A.C., 2006. Effects of Rock Properties on Specific Cutting Energy  
753 in Linear Cutting of Sandstones by Picks. *Rock Mechanics and Rock Engineering*  
754 39. <https://doi.org/10.1007/s00603-005-0062-7>
- 755 Tuncdemir, H., Bilgin, N., Copur, H., Balci, C., 2008. Control of rock cutting efficiency  
756 by muck size. *International Journal of Rock Mechanics and Mining Sciences* 45.  
757 <https://doi.org/10.1016/j.ijrmms.2007.04.010>
- 758 Ucgul, M., Saunders, C., Fielke, J.M., 2018. Comparison of the discrete element and  
759 finite element methods to model the interaction of soil and tool cutting edge.  
760 *Biosystems Engineering* 169, 199–208.  
761 <https://doi.org/10.1016/J.BIOSYSTEMSENG.2018.03.003>
- 762 Wang, X., Wang, Q.-F., Liang, Y.-P., Su, O., Yang, L., 2018. Dominant Cutting  
763 Parameters Affecting the Specific Energy of Selected Sandstones when Using  
764 Conical Picks and the Development of Empirical Prediction Models. *Rock  
765 Mechanics and Rock Engineering* 51. <https://doi.org/10.1007/s00603-018-1522-1>
- 766 Wei, X., Wang, C.Y., Yuan, H.L., Xie, Z.H., 2003. Study of Fracture Mechanism of  
767 Rock Cutting. *Key Engineering Materials* 250.  
768 <https://doi.org/10.4028/www.scientific.net/KEM.250.200>
- 769 Wong, R.H.C., Chau, K.T., 1998. Crack coalescence in a rock-like material containing  
770 two cracks. *International Journal of Rock Mechanics and Mining Sciences* 35, 147–  
771 164. [https://doi.org/10.1016/S0148-9062\(97\)00303-3](https://doi.org/10.1016/S0148-9062(97)00303-3)
- 772 van Wyk, G., Els, D.N.J., Akdogan, G., Bradshaw, S.M., Sacks, N., 2014. Discrete  
773 element simulation of tribological interactions in rock cutting. *International Journal  
774 of Rock Mechanics and Mining Sciences* 65, 8–19.  
775 <https://doi.org/10.1016/J.IJRMMS.2013.10.003>
- 776 Xuefeng, L., Shibo, W., Shirong, G., Malekian, R., Zhixiong, L., 2018. Investigation on  
777 the influence mechanism of rock brittleness on rock fragmentation and cutting  
778 performance by discrete element method. *Measurement* 113, 120–130.  
779 <https://doi.org/10.1016/J.MEASUREMENT.2017.07.043>

- 780 Yadav, S., Saldana, C., Murthy, T.G., 2018. Experimental investigations on deformation  
781 of soft rock during cutting. *International Journal of Rock Mechanics and Mining*  
782 *Sciences* 105, 123–132. <https://doi.org/10.1016/J.IJRMMS.2018.03.003>
- 783 Yurdakul, M., Akdasç, H., 2012. Prediction of specific cutting energy for large diameter  
784 circular saws during natural stone cutting. *International Journal of Rock Mechanics*  
785 *and Mining Sciences* 53, 38–44. <https://doi.org/10.1016/J.IJRMMS.2012.03.008>
- 786 Zeuch, D.H., Finger, J.T., 1985. Rock breakage mechanisms with a PDC cutter, in:  
787 *Conference: Annual Technical Conference and Exposition of the Society of*  
788 *Petroleum Engineers*. Las Vegas.
- 789 Zhou, Y., Zhang, W., Gamwo, I., Lin, J.-S., 2017. Mechanical specific energy versus  
790 depth of cut in rock cutting and drilling. *International Journal of Rock Mechanics*  
791 *and Mining Sciences* 100. <https://doi.org/10.1016/j.ijrmms.2017.11.004>
- 792
- 793