The strength of hard-rock pillars
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Abstract

Observations of pillar failures in Canadian hard-rock mines indicate that the dominant mode of failure is progressive slabbing and spalling. Empirical formulas developed for the stability of hard-rock pillars suggest that the pillar strength is directly related to the pillar width-to-height ratio and that failure is seldom observed in pillars where the width-to-height ratio is greater than 2. Two-dimensional finite element analyses using conventional Hoek–Brown parameters for typical hard-rock pillars (Geological Strength Index of 40, 60 and 80) predicted rib-pillar failure envelopes that did not agree with the empirical pillar-failure envelopes. It is suggested that the conventional Hoek–Brown failure envelopes over predict the strength of hard-rock pillars because the failure process is fundamentally controlled by a cohesion-loss process in which the frictional strength component is not mobilized. Two-dimensional elastic analyses were carried out using the Hoek-Brown brittle parameters which only relies on the cohesive strength of the rock mass. The predicted pillar strength curves were generally found to be in agreement with the observed empirical failure envelopes.

1. Introduction

Pillars can be defined as the in situ rock between two or more underground openings. Hence, all underground mining methods utilize pillars, either temporary or permanent, to safely extract the ore. In coal mines rectangular pillars are often designed in regular arrays such that should a single pillar inadvertently fail the load could be transferred to adjacent pillars causing these to be overloaded. This successive overloading process can lead to an unstable progressive “domino” effect whereby large areas of the mine can collapse. This type of failure occurred in 1960 and resulted in the collapse of 900 pillars in the Coalbrook coal mine in South Africa and the loss of 437 lives. Recently, Salamon [1] summarized the extensive research into coal-pillar design that followed the Coalbrook disaster. The key element that has been used since 1960 for the successful design of coal pillars is “back-calculation”, an approach that has been used extensively in geotechnical engineering [2]. This approach has led to the development of empirical pillar strength formulas but can only be implemented by observing and documenting failed pillars.

The design of hard-rock pillars has not received the same research attention as coal pillar design. This is partly because fewer mines operate at depths sufficient to induce the stresses required to cause hard rocks to fail, and in hard-rock mining pillar and mining geometries are irregular making it difficult to establish actual loads. Nonetheless as mining depth increases the potential for the failure of hard-rock pillars also increases. This paper focuses on the strength of hard-rock pillars, particularly rib-pillars, and presents a stability criterion that can be used to establish hard-rock pillar geometries.

2. Empirical pillar strength formulas

Following the Coalbrook disaster, a major coal-pillar research program was initiated in South Africa. One of the main objectives of this research was to establish the in situ strength of coal pillars. Using the back-calculation approach Salamon and Munro [9] analyzed 125 case histories involving coal-pillar collapse and proposed that the coal-pillar strength could be
adequately determined using the power formula

\[ \sigma_p = K \frac{W^x}{H^\beta} \]  

where \( \sigma_p \) (MPa) is the pillar strength, \( K \) (MPa) is the strength of a unit volume of coal, and \( W \) and \( H \) are the pillar width and height in metres, respectively. The notion that the strength of a rock mass is to a large part controlled by the geometry of the specimen, i.e., the width-to-height ratio, has since been confirmed by extensive laboratory studies, e.g., [10]. The data from the 125 case studies gave the following values for the parameters in Eq. (1): \( K = 7.176 \) MPa, \( x = 0.46 \) and \( \beta = 0.66 \). According to Madden [11] and Salamon [1], Eq. (1) has been applied extensively to the design of pillar layouts in South Africa since its introduction in 1967. While it is tempting to apply Eq. (1) to other pillar designs, it must be remembered that Eq. (1) was developed for room and pillar mining of horizontal coal seams and that the value of \( K \) is only typical for South African coal.

One of the earliest investigations into the design of hard-rock pillars was carried out by Hedley and Grant [3]. They analyzed 28 rib-pillars (3 crushed, 2 partially failed, and 23 stable) in massive quartzites and conglomerates in the Elliot Lake room and pillar uranium mines. These pillars were formed with the long-axis of the rib-pillar parallel to the dip direction of the quartzites. They concluded that Eq. (1) could adequately predict these hard-rock pillar failures but that the parameters needed to be modified to

\[ \sigma_p = K \frac{W^{0.5}}{H^{0.75}} \]  

where the units are the same as Eq. (1). The value of \( K \) in Eq. (2) was initially set as 179 MPa but later reduced to 133 MPa [12].

Since 1972 there have been several additional attempts to establish hard-rock pillar strength formulas, using the “back-calculation” approach (Table 1). Inspection of Table 1 reveals that the best-fit formulas for the observed pillar failures take the form of either a power- or linear-type equation, and that these equations have been used to predict the pillar strength for a wide range of pillar shapes and rock mass strengths as indicated by the unconfined compressive strength (94–240 MPa). Fig. 1 provides a summary of the “failed”, “unstable” and “stable” classes of pillars that were used to establish the pillar strength formulas in Table 1. Fig. 2 shows the predicted pillar strength from the various formulas using a pillar height of 5 m. The

<table>
<thead>
<tr>
<th>Reference</th>
<th>Pillar strength formulas (MPa)</th>
<th>( \sigma_c ) (MPa)</th>
<th>Rock mass</th>
<th>No. of pillars</th>
</tr>
</thead>
<tbody>
<tr>
<td>[3]</td>
<td>( 133 \frac{W^{0.5}}{H^{0.75}} )</td>
<td>230</td>
<td>Quartzites</td>
<td>28</td>
</tr>
<tr>
<td>[4]</td>
<td>( 65 \frac{W^{0.5}}{H^{0.75}} )</td>
<td>94</td>
<td>Metasediments</td>
<td>57</td>
</tr>
<tr>
<td>[5]</td>
<td>( 35.4(0.778 + 0.222 \frac{H}{W}) )</td>
<td>100</td>
<td>Limestone</td>
<td>14</td>
</tr>
<tr>
<td>[6]</td>
<td>( 0.426 \frac{H}{W} )</td>
<td>—</td>
<td>Canadian Shield</td>
<td>23</td>
</tr>
<tr>
<td>[7]</td>
<td>( 74(0.778 + 0.222 \frac{H}{W}) )</td>
<td>240</td>
<td>Limestone/Skarn</td>
<td>9</td>
</tr>
<tr>
<td>[8]</td>
<td>( 0.446(0.68 + 0.52x) )</td>
<td>—</td>
<td>Hard rocks</td>
<td>178</td>
</tr>
</tbody>
</table>

*Database compiled from published sources including those listed in this table.*
pillar strengths in Fig. 2 have been normalized to the laboratory uniaxial compressive strength ($\sigma_c$). As shown in Fig. 2 the formulas predict very similar strengths, particularly for the pillar $W/H$ ratio between 0.5 and 2.5, the range over which all of the pillar failures occur (see Fig. 1).

The stress magnitudes used to establish the pillar strengths formulas in Table 1, which were determined using either the tributary area method, or two and three dimensional elastic analyses, represent either the average maximum pillar stress or the maximum stress at the center of the pillar. In all cases, except the formula presented by Lunder and Pakalnis [8], the pillar-strength formulas ignore the effect of $\sigma_3$ and rely on a simple stress to strength ratio based on the maximum pillar stress and the uniaxial compressive strength. While Lunder and Pakalnis [8] attempt to include the effect of $\sigma_3$ through their parameter $\kappa$ (see Table 1) their formula predicts similar strengths to the other formulas in Table 1 (Fig. 2). Hence the effect of $\sigma_3$ is essentially ignored by the empirical formulas to match the observed failures. This is similar to tunnel stability observations in South African mines where the stability is expressed as a ratio between 0.5 and $\sigma_3$ is essentially ignored by the empirical formulas to match the observed failures. This is similar to tunnel stability observations in South African mines where the stability is expressed as a ratio between 0.5 and 2.5, the range over which all of the pillar failures occur (see Fig. 1).

Fig. 3 illustrates Eq. (3) and shows that the confinement in pillars increases significantly beyond a pillar $W/H$ ratio of 1. Recently, Maybee [14] showed however, that the rate of increase is a function of $k$, the ratio of the far-field horizontal stress $\sigma_1$ and $\sigma_3$ (Fig. 3). Fig. 3 shows that beyond a pillar $W/H$ ratio of 1 the effect of $k$ is significant but for pillar $W/H$ ratios less than 1 the effect of $k$ can be ignored.

The strength of a rock mass is usually described in terms of a constant cohesive component and a normal-stress or confinement-dependent component. Hence for pillars with $W/H$ ratios greater than 1, the strength should increase as the confining stress increases. In the next section the Hoek–Brown failure criterion is used to investigate the effect of confinement on pillar strength.

### 3. Pillar and rock mass strength

One of the most widely used empirical failure criteria is the Hoek–Brown criterion [13]. Since its introduction in 1980 the criterion has been modified several times, most recently in 1997 [15]. The generalized form of the criterion for jointed rock masses is defined by

$$\sigma_1 = \sigma_3 + \sigma_{ci} \left( m_b \frac{\sigma_3}{\sigma_{ci}} + c \right)^a,$$

where $\sigma_1$ and $\sigma_3$ are the maximum and minimum effective stresses at failure, respectively, $m_b$ is the value of the Hoek–Brown constant $m$ for the rock mass, and $c$ and $a$ are constants which depend upon the characteristics of the rock mass, and $\sigma_{ci}$ is the uniaxial compressive strength of the intact rock pieces. For hard-rock masses, Hoek and Brown [15] recommend a value of 0.5 for $a$. In order to use the Hoek–Brown criterion for estimating the strength and deformability of jointed rock masses, “three properties” of the rock mass have to be estimated. These are: (1) the uniaxial compressive strength $\sigma_{ci}$ of the intact rock pieces in the rock mass; (2) the Hoek–Brown constant $m_b$ for these intact rock pieces; and (3) the Geological Strength Index (GSI) for the rock mass. The GSI was introduced by Hoek and Brown [15] to provide a system for estimating the reduction in the rock mass strength for different geological conditions. The GSI can be related to either of the commonly used rock-mass classification systems, e.g., the modified rock-mass quality index $Q'$ defined as

$$Q' = \frac{RQD}{J_a} \times \frac{J_r}{J_a},$$

where $RQD$ is the rock quality designation, $J_a$ is the joint set number, $J_r$ is the joint roughness number, $J_a$ is the joint alteration number or the rock mass rating RMR. Hoek and Brown [15] suggested that GSI can be related to $Q'$ by

$$GSI = 9 \ln Q' + 44$$

![Fig. 3. The increase in confinement at the center of the pillar as a function of k, the ratio of the far-field maximum horizontal stress and vertical stress. The predicted effect of confinement using Eq. (3) is also shown.](image-url)
and to \(RMR\) by
\[ GSI = RMR_{89} - 5, \tag{7} \]
where \(RMR_{89}\) has the Groundwater rating set to 15 and the Adjustment for Joint Orientation set to zero. The parameters \(m_b\) and \(s\) can be derived from \(GSI\) by the following:
\[ m_b = m_i \exp \left( \frac{GSI - 100}{28} \right), \tag{8} \]
\[ s = \exp \left( \frac{GSI - 100}{9} \right). \tag{9} \]

The Elliot Lake uranium orebody was actively mined from the early 1950s through to the mid-1990s. The shallow (10–15°) dipping tabular deposit was characterized by uranium bearing conglomerates separated by massive quartzite beds 3–30 m thick [3,16]. Mining was carried out using room-and-pillar and stope-and-pillar methods with long (76 m) narrow rib-pillars formed in the dip direction. The rock mass quality of the pillars ranged from good to very good (\(Q = 10–100\)) (C. Pritchard, pers. comm.). Seismic surveys carried out across various pillars indicated that at the core of stable pillars the P-wave velocity averaged about 6 km/s while at the edge of the pillars the P-wave velocity dropped to 5.5 km/s [16]. Barton and Grimstad [17] proposed the following correlation between seismic compressional wave velocity and rock mass quality \(Q\) for non-porous rocks:
\[ Q = 10 \frac{V_p - 3500}{1000}, \tag{10} \]
where \(V_p\) is the P-wave velocity in m/s. This relationship is shown in Fig. 4 along with the results from the pillar velocity surveys. These results also support the notion that the pillars were excavated in a very good-quality rock mass. These descriptions and measurements indicate that the rock mass strength can be characterized by a \(GSI\) value of 80 (very good category), which was used to establish the parameters required for the Hoek–Brown failure criterion (Table 2).

<table>
<thead>
<tr>
<th>Parameter</th>
<th>Description/value</th>
</tr>
</thead>
<tbody>
<tr>
<td>Rock-type</td>
<td>Quartzite, Conglomerate</td>
</tr>
<tr>
<td>Insitu stress</td>
<td>(\sigma_1 = 2\sigma_3) and (\sigma_2 = 1.66\sigma_3)</td>
</tr>
<tr>
<td>Intact rock strength</td>
<td>(\sigma_{id} = 230) MPa</td>
</tr>
<tr>
<td>Geological Strength Index</td>
<td>(GSI = 80)</td>
</tr>
<tr>
<td>Hoek–Brown constants</td>
<td>(m_i = 22)</td>
</tr>
<tr>
<td>(m_b)</td>
<td>10.7</td>
</tr>
<tr>
<td>(s)</td>
<td>0.108</td>
</tr>
<tr>
<td>(m_r)</td>
<td>1</td>
</tr>
<tr>
<td>(s_r)</td>
<td>0.001</td>
</tr>
</tbody>
</table>

Fig. 5. Illustration of the suggested post-failure characteristic for a very good quality hard-rock mass.
pillars shown in Fig. 6 are not the actual loads because once failure initiates the loads are redistributed internally within the pillar and/or to adjacent pillars. Numerical analyses were carried out to determine if the pillar strengths predicted using the Hoek–Brown failure criterion with the parameters in Table 2 were similar to the strength predicted by Eq. (2) in Fig. 6. The numerical analyses were carried out using the two-dimensional finite element program Phase2.1 This program has the capability to incorporate the elastic brittle post peak response using the Hoek–Brown parameters. A pillar was considered to have failed when the elements across the pillar had yielded (Fig. 8). This was considered similar to the “crushed” conditions in Fig. 6. However, in order to compare the stress to strength ratio from the numerical program to the data in Fig. 6, the elastic stresses had to be determined for the elastic–brittle failure conditions. These results are presented in Fig. 6 and agree with the failed observations for the pillar \( W/H \) ratio from 0.5 to 1.5. Beyond a pillar \( W/H \) ratio of 1.5 the elastic–brittle response appears to over predict the pillar strength compared to Eq. (2).

4. Pillar stability criterion and GSI

The Phase2 modelling for the Elliot Lake case study used the GSI for a very good-quality rock mass. In the hard-rock mines of the Canadian Shield experience suggests that the GSI will vary significantly. Potvin et al. [6] collected 177 case studies from Canadian hard-rock mines and found that \( Q' \) ranged from 0.1 to 120 (Fig. 9). The GSI values, using Eq. (6) and Potvin et al. [6] database are shown in Fig. 10. The GSI values range from 31 (fair) to 87 (very good) with a mean value of 67, suggesting that the GSI values of 40 (fair), 60 (good) and 80 (very good) would represent the range of typical strength conditions for Canadian hard-rock mines.
corresponding Hoek–Brown parameters for these strength conditions, using Eq. (8) and (9), are given in Table 3. Experience suggests that $m_i = 22$ and $\sigma_d = 230$ MPa are typical values for the hard-rocks found in many Canadian underground mines.

The most extensive database of hard-rock pillar failures was compiled by Lunder and Pakalnis [8] who analyzed 178 case histories from hard-rock mines, 98 of which were located in the Canadian Shield (Fig. 11). Many of these pillars were rib or sill pillars from steeply dipping ore bodies. Lunder and Pakalnis proposed that the pillar strength could be adequately expressed by two factor of safety ($FOS$) lines. Pillars with a $FOS < 1$ fail while those with a $FOS > 1.4$ are stable. The region between $1 < FOS < 1.4$ is referred to as unstable and pillars in this region are prone to spalling and slabbing but have not completely failed, similar to the “partial failure” used by Hedley and Grant [3]. It should be noted that of the pillars investigated 76 were classed as stable; 62 were classed as failed; and 40 were classed as unstable (see Fig. 1). For comparison purposes, the Hedley and Grant pillar strength equation is also shown in Fig. 11. While the Hedley and Grant pillar strength equation is in good agreement with the stability lines proposed by Lunder and Pakalnis [8], Phase2 modelling results using the Hoek–Brown failure criterion for GSI 40, 60 and 80 do not follow the trends of the stability lines proposed by Hedley and Grant [3] or Lunder and Pakalnis [8]. In particular, the slope of the stability lines predicted by the empirical formulas for $W/H < 1$ tend to be somewhat flat and for $W/H = 0.5$ the empirical formulas indicate that the pillar strength ranges from 0.2 to 0.35$\sigma_c$. The slope of GSI lines, however, tend to be steeper and for $W/H = 0.5$ indicate a pillar strength ranging from 0.05 to 0.33$\sigma_c$. The generally steeper slope of the GSI lines reflect the effect of increasing confinement, e.g., see Fig. 3, on the rock mass strength while the observed failure lines appear to be less dependent on confinement. This noticeable trend would suggest that the confining stress-dependent frictional strength component contributes less to the overall pillar strength than the conventional Hoek–Brown failure envelope predicts. However, only with additional case studies can the overall importance of the noted difference in these trend lines be fully assessed.

5. Pillar failure and cohesion loss

The failure of hard-rock pillars involves spalling, i.e., slabbing and fracturing, which leads to the progressive deterioration of the pillar strength. Pritchard and

Table 3

<table>
<thead>
<tr>
<th>GSI</th>
<th>80</th>
<th>60</th>
<th>40</th>
</tr>
</thead>
<tbody>
<tr>
<td>$\sigma_d$ (MPa)</td>
<td>230</td>
<td>230</td>
<td>230</td>
</tr>
<tr>
<td>$m_i$</td>
<td>22</td>
<td>22</td>
<td>22</td>
</tr>
<tr>
<td>$m_s$</td>
<td>10.77</td>
<td>5.27</td>
<td>2.58</td>
</tr>
<tr>
<td>$s$</td>
<td>0.108</td>
<td>0.0117</td>
<td>0.0013</td>
</tr>
<tr>
<td>Residual $m_i$</td>
<td>1</td>
<td>1</td>
<td>1</td>
</tr>
<tr>
<td>$s_c$</td>
<td>0.001</td>
<td>0.001</td>
<td>0.001</td>
</tr>
</tbody>
</table>

develop pillar stability lines based on the rock mass strength. The Hoek–Brown parameters for GSI 40, 60 and 80 given in Table 3 were considered to be representative of the variation of rock mass strength found in Canadian hard-rock mines. The results from this Phase2 modelling are also shown in Fig. 11. While the Hedley and Grant pillar strength equation is in good agreement with the stability lines proposed by Lunder and Pakalnis [8], Phase2 modelling results using the Hoek–Brown failure criterion for GSI 40, 60 and 80 do not follow the trends of the stability lines proposed by Hedley and Grant [3] or Lunder and Pakalnis [8].
Hedley [18] noted that in the early (pre-peak strength) stages of pillar failure at Elliot Lake stress-induced spalling, dominated the failure process while in the latter stages (post-peak strength), after spalling had created the typical hour-glass shape, slip along structural features such as bedding planes and joints played a more significant role in the failure process. These observations are in keeping with the laboratory findings of Hudson et al. [10] and Martin and Chandler [19], who demonstrated that the development of the shear failure plane occurs after the peak strength is reached. Martin [20] proposed that this pre-peak stress-induced spalling/fracturing-type failure is fundamentally a cohesion-loss process and Martin et al. [21] suggested that in order to capture this process in numerical models the Hoek–Brown parameters needed to be modified. They proposed that this spalling- or brittle-type failure could adequately be captured using elastic models and the following Hoek–Brown brittle parameters:

\[
m_b = 0 \quad \text{and} \quad s = 0.11.
\]

The fundamental assumption in using these brittle parameters is that the failure process is dominated by cohesion loss associated with rock mass fracturing, and that the confining stress-dependent frictional strength component can be ignored when considering near surface failure processes. Hence, it is not applicable to conditions where the frictional strength component can be mobilized and dominates the behavior of the rock mass.

A series of elastic numerical analyses were carried out using the boundary element program Examine2D and the Hoek–Brown brittle parameters to evaluate pillar stability over the range of pillar \( W/H \) ratios from 0.5 to 3. The analyses were carried out using a constant \( k \) ratio of 1.5 and the results are presented as solid lines in Fig. 12 for both a factor of safety (FOS) equal to 1 and 1.4. A pillar was considered to have failed when the core of the pillar had a FOS = 1. A similar approach was used to establish when the pillar reached a FOS = 1.4. Fig. 12 shows a better agreement between the FOS lines predicted using the Hoek–Brown brittle parameters, and the FOS lines empirically developed by Lunder and Pakalnis [8] and Hedley and Grant [3] for pillar \( W/H \) ratio less than 1.5. More importantly in contrast to the failure envelopes developed using the Geological Strength Index and the traditional Hoek–Brown parameters (see Fig. 11), the slope of the failure envelope using the Hoek–Brown brittle parameters is in closer agreement with the empirical failure envelopes, particularly for pillar \( W/H \) ratios from 0.5 to 1.5. Also note that for the pillar \( W/H \) ratio less than 1, the strength is essentially constant, reflecting the low confinement for these slender pillars.

Fig. 13 shows the comparison of all the empirical formulas listed in Table 1 with the numerical results using the Hoek–Brown brittle parameters. There is general agreement with the empirical formulas and the predicted pillar strength for \( W/H < 1.5 \) where the majority of the pillar failure occur. Beyond a pillar \( W/H > 2 \) the Hoek–Brown brittle parameters suggest that the strength increases significantly which is in contrast to the empirical formulas. Because at pillar \( W/H > 2 \) the confinement at the core of the pillar is increasing significantly the use of Hoek–Brown brittle parameters will be less appropriate. It should be noted that the pillar-failure database shows that there are only a few pillar failures for pillar \( W/H > 2 \) (Fig. 1). Hence, the empirical pillar strength formulas should be limited to pillar \( W/H < 2 \).
6. Conclusions

Observations [8] of 178 pillar case studies in hard-rock mines indicate that the nearly all failures occur when the pillar $W/H$ ratio is less than 2.5 and that the dominant mode of failure is progressive slabbing and spalling which eventually leads to an hour-glass shape. The Lunder and Pakalnis pillar stability graph documents over 98 rib-pillar observations in Canadian hard-rock mines and is based on the calculated average maximum stress in the pillar, the uniaxial strength of the intact rock and the pillar $W/H$ ratio. Their findings are in keeping with other pillar formulas developed for hard-rock pillars and suggest that for slender pillars ($W/H < 1$) failure initiates at approximately $\frac{1}{3}$ of the laboratory uniaxial compressive strength. For squat pillars ($W/H > 1.5$) these empirical formulas only predict an increase in pillar strength to approximately $\frac{2}{3}$ of the laboratory uniaxial compressive strength, despite a significant increase in confinement at the core of the pillar.

The conventional Hoek–Brown failure envelope is based on a cohesive strength component and a confining stress-dependent frictional component. In a confined state, such as pillar $W/H$ ratios greater than 1, the frictional strength component increases significantly. Two-dimensional finite element analyses using conventional Hoek–Brown parameters for typical hard-rock rib-pillars ($GSI$ of 40, 60 and 80) predicted pillar-failure envelopes that did not agree with the observed empirical failure envelopes. It is suggested that the conventional Hoek–Brown failure envelopes over predict the strength of the hard-rock pillars because the failure process is fundamentally controlled by a cohesion-loss process and for practical proposes the frictional strength component can be ignored at pillar width-to-height ratios less than 1.5.

Two-dimensional elastic analyses were carried out using the Hoek–Brown brittle parameters ($m_{bh} = 0, s = 0.11$). The predicted rib-pillar strength curves were generally found to be in agreement with the observed empirical failure envelopes. It should be noted however, that the Hoek–Brown brittle parameters are not applicable to conditions where the frictional component of the rock-mass strength can be mobilized and dominates the behavior of the rock mass.

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References