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Synopsis

Efficient dynamic bolt reinforcement is crucial in deep-level gold mines prone to rockbursts. Deep-level gold mining in South Africa is approaching 4000 m, so appropriate support designs need to be implemented to manage ground stability challenges associated with these great depths. In order to extract the orebody, both novel mining methods and massive mining techniques have to be applied. Currently, massive mining techniques are used and destressing is imperative. Destressing is accomplished by cutting a series of overlapping, 5 m high horizontal slots across the reef. Although the minor span of each slot is about 100 m, stresses are extremely high due to presence of abutments. In order to protect workers from the effects of small seismic events, a yielding support system consisting of weld mesh (100 mm apertures and 5 mm strands) and 2.5 m long Garford hybrid bolts has been developed. The support system allows installation to be made at 0.5 m from the advancing face without significant damage by the blast. A tendon pretension of 49.3 kN ensures an active support system at installation, with 29 kJ/m² energy absorption capacity during seismic events, and static tensile capacity of 147 kN. The effectiveness of the Garford hybrid bolt is discussed in this paper.

Kevwords

ground support, yielding support, dynamic bolt, Garford bolt.

Introduction

As mining progresses from deep to ultra-deep levels, stress levels increase rapidly. This compromises the effectiveness of support systems and the stability of excavations underground (Ortlepp, 1992). Excavations driven into highly stressed ground typically suffer from stress-induced damage (Mühlhaus, 1990; McCreath and Kaiser, 1992; Maxwell, and Young, 1997; Daehnke et al., 1998). Stress-induced damage can result from either the creation of new fractures or the reactivation of existing fractures in the rock mass (Brady and Brown 1993). As a result of these challenges, rock support systems for underground excavations have changed significantly due to improved technologies and experience gained (Ortlepp, 1992; Jager and Ryder, 1999)

Ground support/reinforcement design

Ground support/reinforcement design has evolved into a comparatively complex discipline involving the quantification of various rock mass and support parameters

(Villaescusa, and Wright, 1997, 1999). A flow chart indicating the principal design steps that should be followed when designing support systems specific to a given geotechnical environment is shown in Figure 1. The generic procedure consists of several distinct steps. In many instances, there may be nothing technically wrong with the design and the performance can be considered acceptable. However, rock mass conditions usually change with the progress of a mine (e.g., stresses increase as depth of mining increases and when the extraction increases) and accordingly, the performance of the ground support may change and become unacceptable (Brady and Brown 1993; Villaescusa, and Wright, 1997, 1999; Jager and Ryder, 1999).

Destress technique

A deep-level gold mine is currently mining at a depth of about 2600 m below surface, with a horizontal-to-vertical stress ratio of about 0.7. Virgin stresses are high in deep to ultra-deep mines, which makes it difficult to implement massive mining techniques. Hence a series of horizontal destress cuts is used to destress the area. The destress cuts are mined through the



Figure 1—The generic procedure of ground support design

* South Deep Goldfields Mine, South Africa.
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strata with a maximum span of about 180 m and a 20 m overlap between successive cuts (see Figure 2). The mining configuration of destress cuts has the shape of an arrowhead to account for proper leads and lags in the high stress conditions. These destress cuts are mined at a stoping height of 5.5 m, the stoping widths are about 5 m, and the lengths 15 m. Subsequently, 10 m of the 15 m is backfilled, to create a 5 m wide strike access drive (SAD), and then the adjacent excavation is mined in the same direction (see Figure 3). In essence, the destressing reduces the vertical stress so that the major principal stress is in the horizontal direction. These destress cuts are supported with 2.5 m long Garford hybrid bolts with 3 m \times 1.5 m rectangular weld mesh on both sidewalls and hangingwall.

The support system in the destress cuts is designed to absorb sufficient energy during a seismic event to protect workers and maintain the integrity of the highly stressed excavations. Hoek and Brown (1980) pointed out that the objective of reinforcing the rock mass is to strengthen it and to prevent loss of strength, thus enabling the rock mass to support itself. The support system within destress cuts also retain the broken rock, which is required for safety reasons. It also becomes essential under high stress conditions as a way of preventing progressive failure leading to unravelling of the rock mass (Zvarivadza *et al.*, 2017).

Methodology

The performance of the Garford hybrid dynamic bolt reinforcement system was evaluated through laboratory tests, which included static pull, dynamic drop, and torque tests), and underground tests (pull test, torque test, and quality of

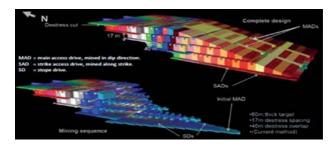


Figure 2—An oblique view of a destress cut

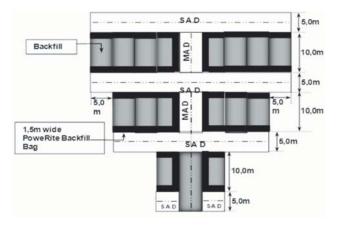


Figure 3—Plan view of horizontal destress layout. MAD – main access drive, SD – strike drive, SAD – strike access drive

the installation). The performance of the bolts during seismic events was also evaluated. It was found necessary to indicate the required support resistance and energy absorption for areas (ground control districts) where Garford bolts were installed.

Support design

The Garford hybrid dynamic bolt reinforcement system was designed to reinforce development ends, also called destress cuts. The previous fallout thickness within the conventional stope and detress cuts, rock mass classification and kinematics analysis, and other rock engineering principles such as rule of thumb were implemented when designing the support requirements for destress cuts. In this case the required support resistance and energy absorption were calculated to determine the required reinforcement system.

Dynamic performance tests

The objective of the dynamic performance tests was to assess the performance of the overall Garford hybrid dynamic bolt system under dynamic loading conditions. Each test consisted of dropping a mass of 3000 kg from a height of 750 mm onto a plate connected to the bolt specimen inside the steel installation tube. The kinetic energy input into the system was 22.06 kJ for each drop test. The testing rig in its actual configuration had a capacity of 3 t; the maximum velocity of 3.26 m/s was used in all tests. The total extension (mm), peak force load (kN), yield deflection (mm), and test time (seconds) were recorded. Figure 4 shows the detailed views of the test assemblies.

Static pull test procedures

All equipment and procedures were developed to meet or exceed the Deep Level Goldmine Standard Test Method for Laboratory Determination of Rock Anchor Capacities by Pull and Drop Tests. Two hydraulic rams with a capacity of 300 kN and total displacement of 300 mm were used. The hydraulic rams were operated, using an electric pump, from the initial phase of loading until the bolt failed, or the rams reached their stroke capacity (300 mm). Figure 5 displays the configuration of the electric pump with Garford bolt. The pressure of the hydraulic ram was measured at both ends of the test specimen (*i.e.* plate and toe ends) using electronic pressure transducers. All measuring instruments were connected to an automatic data-acquisition system and zeroed at the beginning of each test. A data sampling rate of 1 Hz was used.

Underground pull tests and quality control tests

Underground pull tests were conducted by assembling and attaching the pull gear at the collar of the bolts. The dial indicator was attached with the adjustable rod in line with







Figure 4-Drop test assemblies



Figure 5-Static pull test set-up

the bolt axis to allow the bolt to be pulled directly down on the dial indicator, and the dial indicator was adjusted to zero. Immediately after the above steps, loading of the ram from zero to 3 t began and the recorded load and displacement were noted. Continuous pumping and recording of the load for each subsequent 0.05 mm displacement continued until the bolt yielded before full expansion of the ram, when the test was stopped.

Support requirements

A database of all reportable falls of grounds (FOGs) has been kept since 1995. Most of the FOGs occurred in the old conventional stopes and destress cuts, where areal coverage was not good. These records provided valuable information on the support requirements. The reach and capacity of the support was calculated on the 95th percentile of the conventional database (see Figure 6).

Figure 6 shows that the support should anchor above a height of 1.05 m (assuming the suspension theory). An appropriate mechanical anchor length of 1.8 m was chosen to ensure that the loose rock is properly anchored. The support resistance (SR) was calculated assuming a 1.05 m fallout thickness:

$$SR = \rho gt$$
 [1]

where

- ρ is the density of the immediate hangingwall rock (2.68 t/m³)
- *t* is the 95th percentile thickness of measured falls of ground (1.5 m)
- g is acceleration due to gravity (9.81 m/s²).

Factor of safety for steel: 1.2 $SR = (2.68^{\circ}9.81^{\circ}1.5^{\circ}1.2)/(1 \text{ m}^{\circ}1 \text{ m}) = 47.3 \text{ kN/m}^{2}$ (required support resistance).

Brady and Brown 1993, stated that where peak ground velocities exceed 1 m/s, significant kinetic energy and momentum are imparted to the rock, resulting in potentially large displacements. Using the fallout thickness determined above as the potential depth of instability, it is possible to calculate the minimum energy absorption (EA) requirement per square metre of rock wall. The EA criterion for the hangingwall support system at the mine is 30 kJ/m², based on:

$$EA_{hw} = \frac{1}{2}mv^2 + mgh$$
 [2]

where

m is the mass of ejected rock (4020 kg/m²)

v is the anticipated peak ground velocity (3 m/s²)

- h is the distance over which the energy is absorbed (0.3 m) g is acceleration due to gravity (9.81 m/s²).
- EAHW = 0.5*4020*9+(4020*9.81*0.3) = 29.9 kJ (required EA for hangingwall)

EASW = $mgh = 4020^{\circ}9.81^{\circ}0.3 = 11.8kJ$ (required for sidewall).

Support configuration

The support consists of weld mesh, with 5 mm thick strands and an aperture of 100×100 mm, and 2.4 m long yielding anchors. The anchors are installed with 300×280 mm faceplates and pre-tensioned to 30 kN at installation. This tensioning effectively stretches the mesh tight against the hangingwall and provides an active support. The Garford hybrid dynamic bolt spacing is 1 m across and 5 m along the width of the excavation and placed 1.2 m in the direction of the advancing face. The latter spacing fits the 2.4 m face advance. Support configuration in plan view is shown in Figure 7.

Yield support and test results

The Garford hybrid dynamic bolt is a ground reinforcement system that comprises a high-strength steel tube with a slot along its entire length. One end of the friction bolt is tapered to assist insertion into the hole, and the other end has a spherical steel collar to retain a bearing plate and the yielding portion at the end (see Figure 8).

Static pull test results

Five specimens were tested under static conditions to evaluate the installation method. The results obtained are presented as displacements *vs* load curves in Figures 9–13, and summarized in Table I.

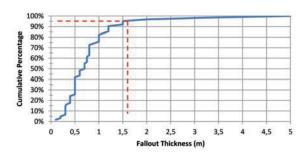


Figure 6—FOG fallout thickness using data from old conventional and destress cuts

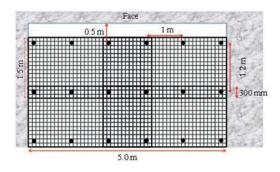


Figure 7-Support configuration in plan view



Figure 8-Schematic view of the Garford hybrid bolt

Table I Summary of static pull test results						
Specimen no.	Peak load (kN)	Sliding load (kN)	Max. extension (mm)			
1	220.634	160	257			
2	232.77	155	270			
3	183.53	165	257			
4	211.88	157	252			
5	186.34	153	280			

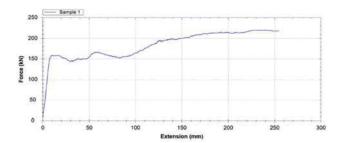


Figure 9-Static pull test for the first specimen

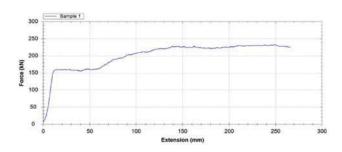


Figure 10-Static pull test for the second specimen

In all tests, after a short sliding distance, the dynamic device anchored into the split set bolt and then the load increased until the solid bolt began to be pulled through the dynamic device. From this test, it can be deduced that the Garford hybrid bolts were capable of withstanding a tensile load ranging from 183.53 kN (18.3 t) to a maximum of 232.77 kN (23.277 t) with a maximum extension of about 280 mm.

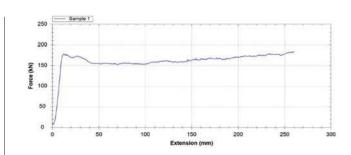


Figure 11-Static pull test for the third specimen

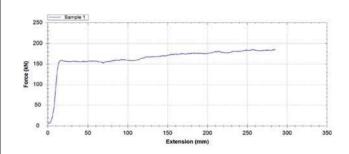


Figure 12-Static pull test for the fourth specimen

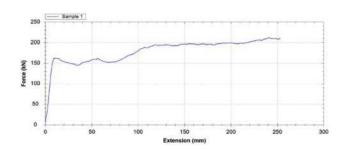


Figure 13-Static pull test for the fifth specimen

Dynamic impact drop test results

The drop test cycle is illustrated schematically in Figure 14. The results from a single drop configuration are summarized in Table II, and the results from tests on five specimens are presented in Figures 15–19. The first specimen withstood the first impact with a peak load of 191.79 kN, and the time

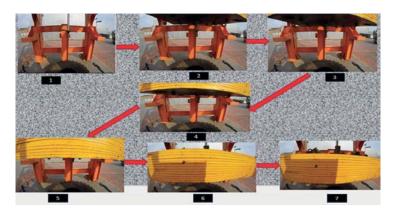


Figure 14-Drop test cycle on a Garford hybrid bolt

Table II										
Dynamic impact drop test results with the continuous tube configuration										
Sample	Drop height (mm)	Energy (kJ)	Max. velocity (m/s)	Peak force (kN)	Test time (s)	Total extension (mm)	Yield deflection (mm)	Peak force between 120–250 kN	Energy between 21–26 kJ	Yield deflection between 200–300 mm
1	750.0	22.06	3.26	191.79	20.20	971.37	221.00	Pass	Pass	Pass
2	751.0	22.09	2.90	217.95	21.10	941.54	191.00	Pass	Pass	Pass
3	752.00	22.12	3.28	141.45	21.40	980.44	228.00	Pass	Pass	Pass
4	755.00	22.21	3.28	138.18	17.50	986.85	232.00	Pass	Pass	Pass
5	750.00	22.06	3.25	157.98	20.50	956.46	206.00	Pass	Pass	Pass

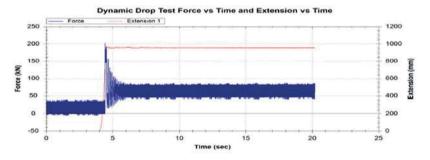


Figure 15-Drop test on the first specimen

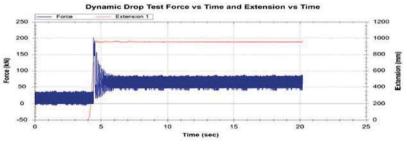


Figure 16-Drop test on the second specimen

taken for a test was 20.20 seconds, with a yield deflection of 221 mm. During this test, a maximum velocity of 3.26 m/s was achieved, with the energy absorbed being 22.06 kJ. After the test, no longitudinal cracks were observed along the bolts, unlike other types of bolt. The second specimen

withstood the impact with a peak load of 217.95 kN and the time taken for a test was 21.10 seconds, with a yield deflection of 191mm. A maximum velocity of 2.9 0m/s was achieved with the energy absorbed being 22.09 kJ. Unlike the first and second tests, the third specimen withstood the

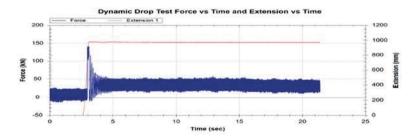


Figure 17-Drop test on the third specimen

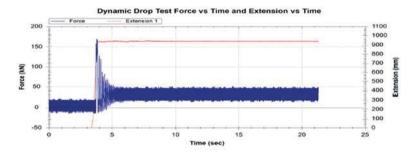


Figure 18-Drop test on the fourth specimen

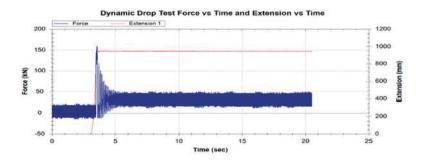


Figure 19-Drop test on the fifth specimen

impact with a peak load of 141.45 kN and the time taken for a test was 21.40 seconds, with a yield deflection of 228 mm. The maximum velocity was 3.28 m/s, with the energy absorbed being 22.12 kJ. The fourth specimen withstood the impact with a peak load of 138.18 kN and the time taken for the test was 17.50 seconds, with a yield deflection of 23 2 mm. During this test, a maximum velocity of 3.2 8m/s was achieved with 22.21 kJ energy absorbed. The last specimen was found to withstand the impact with a peak load of 157.98 kN and the time taken for a test was 20.50 seconds, with a yield deflection of 206 mm. The maximum velocity was 3.25 m/s and the energy absorbed 22.06 kJ. Figure 14 confirms the sliding of the dynamic device into the split set before plastic deformation of the solid bolt.

Torque tests

294

The laboratory torque test was conducted using nuts in different conditions, including rusted, new, and gritty nuts. The performance of all nuts was found to be effective and suitable for the bolts. Figure 20 shows the torque test performance on surface.

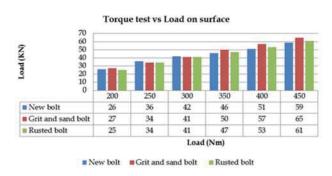


Figure 20-Garford bolt torque tests vs load (tests on surface)

Underground pull tests

To investigate the behaviour of Garford bolts in actual mine conditions, underground pull tests were conducted. The results of the study indicated that, as expected from the surface test results, none of the tested bolts could withstand more than $160~\rm kN$, most of the bolts failing between $60~\rm kN$

and 130 kN (see Figure 21). Only one bolt withstood 160 kN. Further underground investigations included torque tests and assessing the quality of installation.

Performance of Garford hybrid bolts under dynamic conditions

During the trials, several seismic events occurred ahead of the mining faces and in the back areas. Most of the areas affected by these events were found to be stable and the performance of the bolts was found to be effective. Although

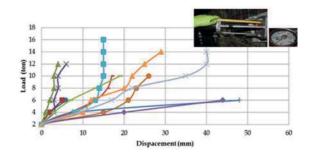


Figure 21 — Underground pull test results for Garford hybrid dynamic bolts

some of the bolts were noted to pull out from the in-stope pillar, the majority of the bolts managed to withstand several seismic events (see Figures 22–24).

Underground torque test results

Underground torque tests were conducted using a torque tester. The tests were assessed based on the number of turns



Figure 24-Performance of Garford bolt in a seismic event

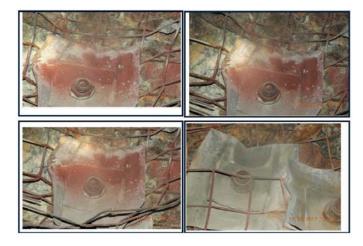


Figure 22-Performance of Garford bolts in a seismic event

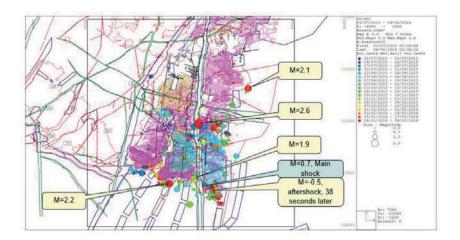


Figure 23-Localities of seismic events

Table III

Torque test results

Garford hybrid dynamic bolt torque tests								
Bolt number	Before test	After test	Turns	Bolts number	Before test	After test	Turns	
1	Tight	2600 Nm	No turn	10	Tight	2600 Nm	11/2 turn	
2	Tight	2600 Nm	No turn	11	Tight	2600 Nm	No turn	
3	Tight	2600 Nm	11/2 turn	12	Tight	2600 Nm	No turn	
4	Tight	2600 Nm	No turn	13	Tight	2600 Nm	No turn	
5	Tight	2600 Nm	11/2 turn	14	Tight	2600 Nm	No turn	
6	Tight	2600 Nm	No turn	15	Tight	2600 Nm	11/2 turn	
7	Tight	2600 Nm	No turn	16	Tight	2600 Nm	No turn	
В	Tight	2600 Nm	No turn	17	Tight	2600 Nm	11/2 turn	
9	Tight	2600 Nm	No turn				_	

in which the span rotated. The results of the study showed that five of the tested bolts out of seventy were not properly tensioned (the turns were either a half turn, full turn, or quarter turn) (see Table III).

Underground performance

The actual performance of the bolts underground was found to be different from that in surface tests. Several contributing factors to this were identified, including poor angle of installation, poor tensioning, the use of a larger drill bit than required, and highly fractured ground conditions which hampered proper installation. Poor support spacing was also found to contribute to the ineffectiveness of the Garford bolt in some area.

These issues have, to some extent, been addressed by providing sections with dedicated air compressors and hydraulic drilling machines. The bit size has been reduced to 41 mm to make it easily differentiable from the 45 mm blasthole bit. It is also possible to fit standard drill-rod couplings into the 41 mm diameter support holes.

During the trials, several seismic events occurred ahead of the mining faces and also in the back areas. It was noted that most of the areas affected by events were found to be stable and the performance of the bolts were found to be effective. Although some of the bolts were noted to pull out from the in-stope pillar, majority of the bolts manage to withstand several seismic events (see Figure 23 and 24).

Conclusions

In order to determine the dynamic capability of the Garford hybrid bolt, static and dynamic tests were performed at the Videx Mining test facility in South Africa. The Garford hybrid bolt was found to withstand a maximum peak load of 217.9 5kN, with a minimum peak load of 138.18 kN; this was achieved in a test time of 17.50-21.40 seconds. Energy absorption ranged from 22.06 kJ to 22.21 kJ, with the velocities ranging from 2.90 m/s to 3.28 m/s. The maximum yield deflection of the bolts was found to range from 191 mm to 232 mm, indicating that the bolts were able to stretch adequately. The tensile tests have shown that the Garford hybrid bolts were capable of withstanding a tensile load ranging from 183.53 kN (18.3 t) to a maximum of 232.77 kN (23.277 t), with a maximum extension of about 280 mm. The underground performance of the bolts was different from that in the surface tests, but this was largely due to the quality of installation and ground conditions. Garford hybrid bolts were

found to perform well during seismic events that occur at the vicinity of the mining faces and back areas.

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