P-03-74

Oskarshamn site investigation

Q-logging of KSH 01A and 01B core

Nick Barton, Nick Barton & Associates

July 2003

Svensk Kärnbränslehantering AB

Swedish Nuclear Fuel and Waste Management Co Box 5864 SE-102 40 Stockholm Sweden Tel 08-459 84 00 +46 8 459 84 00 Fax 08-661 57 19 +46 8 661 57 19



ISSN 1651-4416 SKB P-03-74

Oskarshamn site investigation Q-logging of KSH 01A and 01B core

Nick Barton, Nick Barton & Associates

July 2003

This report concerns a study which was conducted for SKB. The conclusions and viewpoints presented in the report are those of the author and do not necessarily coincide with those of the client.

A pdf version of this document can be downloaded from www.skb.se

Summary

The first Simpevarp potential repository site borehole KSH 01A and 01B cores provided core from 1.3 to 1003.0 m depth. This was independently Q-logged by NB&A during a three-day period (12th-14th May, 2003), without access to BOREMAP results or regional jointing frequencies or orientations. The Q-logging was intended to be an independent check for subsequent BOREMAP-derived Q-parameter estimation.

The Q-logging was accomplished using the manually-recorded 'histogram method' which allows the logger to enter Q-parameter ranges and depths directly into the appropriate histograms, which facilitates subsequent data processing using Excel spreadsheets. Successive pairs of core boxes, which contain an average of 11 meters of core in ten rows, were the source of ten opinions of each of the six Q-parameters, giving a total of approximately 5400 recordings of Q-parameter values for the 180 core boxes.

Data processing was divided into several parts, with successively increasing detail. The report therefore contains Q-histograms for the whole core, for the seven identified fracture(d) zones combined as if one unit, and then for the whole core minus these fracture(d) zones. This background rock mass quality is subsequently divided into ten depth zones or slices, and trends of variation with depth are tabulated. From the seven identified fracture(d) zones, four principal ones are selected and analysed separately, and similarities and subtle differences are discerned between them.

The overall quality of this first core is 'good', with Q(mean) of 12.5, and a most frequent Q-value of 29, but there is significant jointing until about 700 m depth. The *typical* range of quality is from 0.3 to 600, which covers most of the upper half of the six order of magnitude Q scale. The fracture(d) zones, representing some 9% of the 1002 m cored, have a combined Q(mean) of only 1.1 ('poor') and a range of *typical* quality of 0.02 to 21, or 'extremely poor' to 'good'.

Contents

1	Introduction	7
2	Q-logging methodology	9
3	Examples of joint character	11
4	Overall quality of KSH 01A and 01B core	15
5	Character of fracture(d) zones	17
6	Character of KSH 01A and 01B, minus the fracture(d) zones	19
7	Individual character of fracture(d) zones	21
8	General variation with depth (minus fractured zones)	27
9	Conclusions	33
10	Recommendations	35
Refe	erence	36
Арр	endix A – Histograms	37
Арр	endix B – The Q-method of rock classification	49

1 Introduction

The writer performed Q-logging of 1002 m of core from the first site characterization borehole(s) KSH 01A and 01B between 12th and 14th May. This work was requested by Rolf Christiansson, for the purpose of supplementing the more detailed BOREMAP geological logging.

It is intended that this Q-logging can provide an independent check of BOREMAPderived Q-parameter data, which is under preparation following geological logging of this first deep borehole-core.

2 Q-logging methodology

It was the intention of SKB that this Q-logging should be of an 'overview' character. For this reason time was limited, and the 1002 m approx of core (1.25 to 100.8 from KSH 01B, and the remaining 902 m from KSH 01A) was Q-logged using the 'histogram method' /Barton et al, 1992/. The 1002 m was logged in about 20 hours.

The procedure used was to log two core boxes at a time. Due to the 1.1 m approx length of the boxes, there were a total of 87 pairs of core boxes, giving close to 11 m of core on average.

The Q-histogram logging method, described recently in some detail by /Barton, 2002/, consists of making estimates of the variability of each of the six Q-parameters. Each of the Q-parameters is defined, and complete ratings listed, in Appendix B at the end of this report.

For each pair of core boxes, imagining there was the normal 1.0 m of core length, a total of 10 opinions were recorded concerning the visible variability of each of the six Q-parameters. In many cases, such as RQD=100% in excellent rock, there was of course little variation, and logging could proceed much faster. *In Appendix A, scanned copies of the ten hand-filled Q-bistogram logging sheets will be found.*

These will be seen to contain numerous entries 111111, 22222, 666, 777, 999999 etc in the appropriate boxes. Each number, from 1 through 9 (sometimes to 10), is related to a specific core depth, as listed in the left margin of each sheet. Each number is also placed in the Q-parameter box appropriate to the observed/estimated quality (or lack of quality, as the case may be). For 1002 m of core, with 10 opinions of each of the six Q-parameters, there were a resulting total of 10x6x87 or more than five thousand Q-parameter estimates.

The result is overall histograms of variability (or similarity at deeper levels in the rock mass) plus the depth-related variability from which depth logs can be extracted if desired.

Emphasis here has been to *characterize* the overall variability of the rock mass, especially of the different joint sets, as opposed to a specific tunnel-related *classification* for estimating rock reinforcement and support needs. In the latter, the least favourable J_r/J_a ratio is considered, together with the tangential stress effect of the tunnel. The term SRF is evaluated considering the ratio σ_c/σ_1 when support design is the objective. When excavations are considered at this depth, there may be a significant reduction in Q-value to the *Q-classification* value, due to low potential σ_c/σ_1 ratios and elevated SRF values, in particular when the rock is massive, and the 'relative block size' (represented by RQD/J_n) is significantly less than say 25.

The initial purpose of the *characterization* performed, will be to apply empirical linkages between Q-values (more specifically Q_c values) and engineering parameters such as deformation modulus, cohesion, friction and uniaxial rock mass strengths – to the extent that these 'continuum' concepts apply.

For these empirical conversions, the variation of uniaxial strength for the different rock types encountered down the core will be required, as Q_c is calculated from the product of Q and the normalized (by 100 MPa) ratio of $\sigma_c/100$. Estimates from Schmidt hammer recordings, for extrapolating the concentrated sets of laboratory data, should be sufficiently accurate for this exercise.

3 Examples of joint character

Local road cuttings in the area of the nuclear power plant, such as illustrated in Figure 3-1, suggest moderate, steep and very steep dip for three or more joint sets in the monzonitic and vulcanite rock types, which show considerable variation through much of the core.

During the Q-logging of the core, joint roughness traces of representative examples of each joint set were recorded. Some selected examples are shown in Figures 3-2 and 3-3. Here we see both J_r and J_a values, due to the presence of clay coatings or thin clay fillings, which have powdered and fallen into the core boxes. We see examples of $J_r = 1, 1.5, 2(-)$, and 3 for the moderate, steep and very steeply dipping joints.

There are several examples of vertical joints of considerable roughness ($J_r = 3$ to 3+) in the first 150 m of the core, but these are seldom in evidence elsewhere, and extra account of them in the form of possible hidden 'random' features has not been added to J_n , except where they are actually seen in the core.

Due to the dominance of dipping structures (as opposed to vertical structures) it can probably be assumed that vertical boreholes are good samplers of the main jointing at Simpevarp, at least in the nuclear power plant locality where the first two holes are drilled.



Figure 3-1. Examples of joint character and orientation in local exposures.



Figure 3-2. Examples of $\mathcal{J}_r = 1.5$ (or 2.0–) and $\mathcal{J}_a = 4$ due to clay coating/thin filling.



Figure 3-3. a) Examples of $\mathcal{J}_r = 1.0$ and 1.5. b) Example of $\mathcal{J}_r = 3$ and $\mathcal{J}_a = 4$.

4 Overall quality of KSH 01A and 01B core

The first procedure of Q-histogram *analysis* was to count all recordings of quality from the ten logging sheets in Appendix A, including those of obvious fracture(d) zones, and produce Q-parameter histograms for the complete 1002 m of core to 1003.0 m depth. The result is shown in Figure 4-1. This is derived from the summation of the ten sheets of recordings, shown, also hand-recorded, in sheet 11 in Appendix A.

The most frequent quality is 'good' with $(Q_{most frequent}) = 29.3$. The weighted mean (weighted downwards by at least seven distinct fracture(d) zones) shows $Q_{mean} = 12.5$ (also described as 'good'). *Typical* minimum and maximum values range from about 0.3 to 600 – 'very poor' to 'exceptionally good'.

The long 'tail' on the RQD distribution, together with the dominance of three joint sets in the widely spread J_n distribution, indicates the generally jointed nature of much of the core. As will be seen in the subsequent quality-depth statistics, there are nevertheless distinct variations down the core, with particularly jointed sections in the 200 to 300 m and 400 to 500 m depth zones. Beyond 700 m the quality is consistently very good.



Figure 4-1. Q-parameter bistograms for the complete KSH 01A and B cores, from 1.3 to 1003.0 m depth.

5 Character of fracture(d) zones

Inspection of the character and distribution of individual Q-parameters – particularly lower-valued 'tails' of RQD, and higher-valued 'tails' of J_n and J_a – give a strong indication of fracture zones, which subsequently may receive the tentative notation *fracture zone*. The Q-parameter histograms for these zones show lower quality tails in the distribution that all trend to the left. Higher qualities trend only to the right. There are both skewed distributions (e.g. RQD = 100% dominating), and more normal distributions (i.e. $J_n = 2$ to 4 dominating).

In the present report *seven* zones of noticeably increased fracturing have been identified, where presumably both the BOREMAP geologists and certainly the Q-logger had to take more time due to all the details of jointing and fracturing to be recorded. The present Q-parameter based identification of fracture(d) zones, which is entirely independent of the geological logging assessment (whose result is unknown to the undersigned), is as follows:

FZ 1	depth 138.5 to 154.5 m (approx)	sheet 2, ref 5,6	Appendix A
FZ 2	depth 247.7 to 253.0 m (approx)	sheet 3, ref 6	Appendix A
FZ 3	depth 407.0 to 409.5 m (approx)	sheet 5, ref 1	Appendix A
FZ 4	depth 420 to 437 m (approx)	sheet 5, ref 3,4	Appendix A
FZ 5	depth 541 to 570 m (approx)	sheet 6, ref 5,6,7,8	Appendix A
FZ 6	depth 619 to 637 m (approx)	sheet 7, ref 2,3,4	Appendix A
FZ 7	depth 725 to 730 m (approx)	sheet 8, ref 4	Appendix A

NOTE: There are in addition two small regions of 'crushed core' at 829.5 to 830.0 (approx) and at 901.0 to 902.0 m (approx). These may be due to damage during core recovery according to BOREMAP personnel.

The above seven 'fracture(d) zones' have first been assembled as a typical 'unit' (combining the characteristics of all seven zones) prior to individual histogram representation, which obviously is more correct. The preliminary combined result is shown in Figure 5-1

The summary statistic of the seven combined FZ zones is as follows, giving immediately the (correct) impression that the rock mass quality is of distinctly lesser quality than the generally good quality of the remainder.

The very wide distribution of RQD seen in Figure 5-1 shows that the joint spacing varies widely in these fracture(d) zones, but the dominance of at least three joint sets ($J_n = 12$) and a considerable statistic of thinly filled and coated and weathered joints, suggests also a variable but sometimes rather permeable condition.



Figure 5-1. Q-parameter histograms for the seven identified fracture(d) zones.

6 Character of KSH 01A and 01B, minus the fracture(d) zones

By counting overall Q-parameter observation totals (10x6x87 = 5220 (approx) in Appendix A) and *subtracting* the fracture(d) zone recordings listed in Section 5 (about 490 observations for the seven zones), we obtain the 'net result' for the rock mass minus the fracture(d) zones. The above numbers suggest that about 93 m/1002 m = 9.3% is significantly fractured, *as measured in a down-hole direction*. If, as may be assumed, some of the zones have significant dip angles, then this percentage would be reduced with respect to perpendicular measurement.

The 'net or back-ground rock mass' result is shown in Figure 6-1, and demonstrates a generally 'good' to 'very good' quality. The following Q statistics can be noted:

Perhaps surprisingly, there is still a wide distribution of J_n , in other words a marked number of joint sets, despite the subtraction of the seven identified fracture(d) zones. There is also still a significant 'tail' of RQD out to medium values of 60 or 70, but the extreme low values of RQD have of course been almost 'eliminated' by this subtraction of marked fractured zones.



Figure 6-1. Q-parameter histograms for the back-ground rock mass, with the seven fracture(d) zones excluded.

7 Individual character of fracture(d) zones

In this section, possible differences in character of the four principal fracture(d) zones are investigated, which may be useful when subsequent deep boreholes are compared and 'cross-correlated' – if this proves possible. Figures 7-1, 7-2, 7-3 and 7-4 show individual Q-parameter histograms of the widest of the presently identified fracture(d) zones, which from Section 5 are seen to be FZ 1, FZ 4, FZ 5 and FZ 6. These have a combined length down the borehole of approximately 16+17+29+18 m respectively, so dominate as 80 m of the 93 m total of presently delineated zones in this Q-parameter based report.

Two photographic examples of these fracture(d) zones, taken from FZ 4 and FZ 5 are shown in Figure 7-5. These boxes are from 421.1 to 431.3 m, and 557.8 to 567.9 m respectively. In terms of tunnel stability, these of course would not be serious occurrences, apart from increased inflows – or the possible need of pre-injection.

Comparison of Figures 7-1 through 7-4 shows that each of the fracture(d) zones has bimodal distributions of RQD. In other words, despite the lower general quality, they have a 'core', or several locations, of significantly worse quality. Nevertheless, the shallowest zone FZ 1 has a significantly less 'crushed' appearance than the deeper zones, with higher overall RQD distributions.

Three joint sets and random ($J_n = 12$) dominate in the two deepest zones FZ 5 and FZ 6, but not in the two shallower zones. Altered or stained joints, clay coatings and thin, sometimes thicker fillings are seen in all four zones, i.e. there are frequent values of $J_a = 2, 3, 4$ and 6. In terms of Q-value statistics, there are remarkable similarities in the *mean* and *most frequent* values for the four principal zones, as seen in the following table.

These identified fracture(d) zones are easy to see when surveying the core, and presumably will have resulted in lower terrain if intersecting the ground surface. They represent significant reductions in quality, and perhaps would result in at least 1 km/s reduction in P-wave velocity in relation to 'background' velocities. This may well be the reason for the (moderate) height differences in the local ground surface.

Fracture(d) zone	Depth (m)	Q _{most frequent}	Q _{mean}	Q _{typ. min.}	Q _{typ.max.}
FZ 1	138.5–154.5	5.3	2.2	0.3	7.0
FZ 4	420-437	3.8	1.4	0.02	24.8
FZ 5	541-570	2.3	1.3	0.02	20.9
FZ 6	619-637	2.7	1.1	0.04	10.5

Table 7-1. Q-statistics for the four principal fracture(d) zones identified from Q-parameter changes.



Figure 7-1. Individual character of FZ 1 (138.5–154.5 m).



Figure 7-2. Individual character of FZ 4 (420–437 m).



Figure 7-3. Individual character of FZ 5 (541–570 m).



Figure 7-4. Individual character of FZ 6 (619–637 m).



Figure 7-5. Examples from FZ 4 and FZ 5.

8 General variation with depth (minus fractured zones)

Since the fracture(d) zones have been analysed in some detail above, it is logical to finally separate them from the remaining 91% (approx) of the better quality core, and investigate if there are significant trends of variation in the 'background rock quality' with depth. This can be done at this stage only in relation to the Q-logging. Geological variation, and rock type changes (i.e. also potential strength changes) cannot be evaluated at this stage.

The procedure adopted to extract the required 'background rock mass quality' data, was to take each Q-logging sheet in turn (approx 100 m of core per sheet, see Appendix A) and subtract the Q-parameter recordings of the seven identified fracture(d) zones as appropriate.

The results of key Q-value statistics for the ten '100 m thick' slices down the borehole are presented in Table 8-1. Each logging sheet represents a maximum of about 100 m of core, from which the seven fracture(d) zones are subtracted as they occur. This means that one of the ten 'slices' is reduced to only about 70 m in (down-hole measured) thickness, due to the maximum 29 m length of identified fracture(d) zone FZ 5.

Table 8-1 reflects the general improvement of 'back-ground' rock from 500 m depth and beyond. It may also be emphasised that beyond 700 m depth – the last three 100 m slices – there is only one small fracture(d) zone (FZ 7 of 5 m thickness).

Depth down hole	Q _{most frequent}	Q _{mean}	Q _{typ. min.}	Q _{typ. max.}
1.3–100.8 m	24.8	6.3	0.9	75
100.8–199.9 m	31.4	12.9	2.6	100
199.9–302.9 m	13.9	6.0	0.3	75
302.9–398.8 m	31.4	16.8	1.8	150
398.8-500.1 m	15.8	15.3	2.7	150
500.1-600.6 m	49.5	17.9	4.0	200
600.6-699.5 m	62.7	14.0	2.3	200
699.5-798.8 m	133	48.3	2.6	600
798.8-901.2 m	133	59.4	7.1	300
901.2-1003.0 m	150	71.7	13.9	1200

Table 8-1.	Variations	with depth	for the '	'background	rock mass'	(minus 7	x FZ).
------------	------------	------------	-----------	-------------	------------	----------	--------

The typical appearance of two parts of the 'background rock mass' is shown in Figure 8-1. The depths shown are (top) 302.9 to 313.4 m (moderately jointed), and (bottom) 983.0 to 994.1 m (massive). The local Q-parameters were estimated to be within the following ranges for these two 10 m slices of the rock mass:

1. (above) $Q = (80-100)/4 \ge 2/1 \ge 0.66/0.5 = 53-66$ ('very good'),

2. (below) Q = $100/1 \ge (2-4)/1 \ge 1/0.5 = 400-800$ ('exceptionally good').

There is considerable uncertainty about the most massive rock, as joint continuity cannot be assessed, and J_r has also been given the possible value of 4, which represents 'discontinuous'. If the rock mass was relatively free of joints for distances of many tens of meters, the effective Q-value as regards rock mass parameters appropriate to engineering-scale problems is likely to be greater than 1000, as reflected by the Q typical maximum value of 1200 given in the last row of Table 8-1.

The Q-parameter histograms for the three deepest '100 m slices' are finally given in Figures 8-2, 8-3 and 8-4. This is the best quality region, based on a sum of 11 out of 12 of the top-score Q-value criteria (i.e. $3x Q_{most frequent}$, $3x Q_{mean}$, $2x Q_{typical minimum}$ and $3x Q_{typical maximum}$), as can be seen by checking Table 8-1.



Figure 8-1. Examples of jointed and massive 'back-ground' rock mass.



Figure 8-2. Q-bistograms for 700-800 m (approx).



Figure 8-3. Q-bistograms for 800–900 m (approx).



Figure 8-4. Q-bistograms for 900–1000 m (approx).

9 Conclusions

- 1. Q-logging using the histogram method is found to be an efficient way of collecting the extensive range of rock mass characteristics represented in a little over 1000 m of core. Eighty seven pairs of core boxes, each containing about 11 m of core, were logged with ten allowable opinions of the local rock mass conditions (+/- a few meters). Since there are six Q-parameters, this data set consists of 10x6x87 = 5,220 observations. Concerning RQD, the ten data per pair of core boxes related to each 1.1 m length of core.
- 2. The hand-recorded data, giving depth and joint character in each box of the histograms, required ten data sheets, one for each 100 m of core. This was processed in Excel spreadsheet format. The data was initially divided into three parts, namely the whole 1002 m of core, the seven identified fracture(d) zones termed FZ 1 to 7, and the whole core minus the seven principal fracture(d) zones.
- 3. The whole core displayed 'good' quality, with $Q_{mean} = 12.5$, and $Q_{most frequent}$ equal to 29. The typical range of Q-values for the whole core was about 0.3 to 600, the latter based on the assumption of virtually no jointing in the lowest portions of the hole. The seven identified fracture(d) zones, if treated as one unit, showed Q_{mean} and Q_{most} frequent as low as 1.1 and 2.3 respectively (i.e. 'poor'). They also displayed, collectively, the complete range of RQD from zero to 100%. These fracture(d) zones constituted some 9% of the core, and when excluded, the remaining 91% or 910 m of core showed $Q_{mean} = 16.2$.
- 4. Individual histogram treatment of four of the principal fracture(d) zones revealed that they each displayed bimodal distributions of RQD. In other words, despite lower general quality, they also each had a 'core', or several locations, of significantly more jointed or fractured conditions. Three joint sets plus random dominated in the two deepest zones FZ 5 and FZ 6, and relative mean block size ratios (RQD/J_n) of the four selected zones were only 7.8, 3.8, 4.9 and 5.3, signifying small average block sizes, with typical minima on either side of 1.0 (e.g. 10/15, and 25/15).
- 5. The whole core mean value of J_w was 0.75, and for the seven fracture(d) zones only 0.52, implying 'high pressure inflows' in the tunnelling context i.e. the possible need for pre-injection of such zones. The 91% of 'background rock mass' (with seven fractured zones excluded) had a mean estimated J_w of 0.77, which is little improved due to the general level of jointing, and therefore potential connectivity, in the 'background rock mass'.
- 6. There is a recent Q-logging 'footnote' /Barton, 2002/ relevant for *characterization* (distant from excavations) for potential reduction of J_w with successive depth zones. Since RQD/J_n values were easily as low as the stipulated range (0.5 to 25) and often in the range 10 to 15, implying good connectivity, J_w values as low as 0.5 and even 0.33 were applied where this seemed appropriate due to intersecting, frequent jointing. There could be exceptions to this in portions of fracture(d) zones containing clay fillings.

7. An overall comparison of the whole 1002 m of core, and about 90 m total of fracture(d) zones, reveals weighted mean values of RQD/J_n (relative block size) and J_r/J_a (friction coefficient) declining from whole sample (1002 m) values of 88/5.9 and 1.8/2.1, to only 54/10.8 and 1.4/3.9 for the fractured zones. The implied reductions in block size and shear strength are quite significant.

10 Recommendations

For the empirical Q-system correlations to rock mass properties that are to be assessed in later reports, it is essential to have some level of knowledge of rock strength variation down the length of the core, so that the presently reported Q-value variations can be converted to Q_c values, which forms the main basis of full characterization and rock mass property estimation (where $Q_c = Q \ge \sigma_c/100$). It should be sufficient to utilize estimates of σ_c from point load or even Schmidt hammer testing, once reliable site specific correlations to σ_c are determined or agreed.

Reference

Barton N, Løsel F, Smallwood A, Vik G, Rawlings C, Chryssanthakis P, Hansteen H, Ireland T, 1992. Geotechnical Core Characterisation for the UK Radioactive Waste Repository Design. 1992 Proc. of ISRM Symp. EROCK, Chester, UK.

Barton N, 2002. Some new Q-value correlations to assist in site characterization and tunnel design. Int. J. Rock Mech. & Min. Sci. 39, 185–216, Pergamon.

- 1. Ten hand-filled Q-histogram logging sheets containing the raw data from which subsequent EXCEL calculations were performed.
- 2. One hand-filled Q-histogram of total numbers-of-observations for the ten '100m slices', with exact depths given on the left.

(A)	Locat	ION: SIMPE	VARP		Depth / cha	ainage: /D0.g		Date: /3 Page: /	1/5/03
Numbers	Q (ty	pical range) =			mean) =		Q (most t	(peri	
domains, core boxes, tunnel lengths (under- line, or spacify)	влоск		20 30	40 5	5 1/2 0 60	9 6 6 70 80	Good 12 35 1 12 35 1 12 35 1 14 12 35 1 15 55 4 15	Ext 7777	RQD % sre piece 10 cm
1=1-25- 11-92 2=11-82-	SINES	Earth Foar	4 4 4 4 5 5 7 6 7 5 7 6 7 5 7 6 7 5 7 6 7 5 7 6 7 7 6 7 7 6 7 7 6 7 7 7 6 7 7 7 7	ree 410 9 8 2 82 6 6 2 1 82 6 7 6 7 6 7 82 7	Two 277 2 8828 66727 5666 144569 332255 11/2122 4 3	One of the other states of	None	N	J _a lumber o int sets
22-40 = 22-6- 33-1 = 33-1- 64-0 = 64-0 = 54-9 = 51-9-	FR-0	-Fils_	B	1001 1014	Unduli 1771 82.0 7791 82.0 7791 85.64 14.55 44.93	41mg 0 8 44 12 9 9 8 44 14 9 9 9 8 44 14 9 9 9 8 44 14 9 9 9 9 8 44 14 9 9 9 9 9 8 44 14 9 9 9 9 9 9 8 44 14 9 9 9 9 9 9 8 44 14 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9	086.	ji re ta	J, oint iughnesa ieast ivourable
65-8 =(5-8- ++-1 = 77-1- \$77-5 = 87-5- = 87-5- = 87-5- = 87-5-	T-ON	20 13 12 10	8 6 5	1.3 Thin 12.8		2201ed 14 14 15 14 15 14 14 15 14 14 14 14 14 14 14 14 14 14	Unfiled)	June 1	J _a bint teration least rvourable
or Sketch +98-S- 100-3 END	ACT-VE ST	Esc. Info 0.05 0.1 Squeeze	0.2 0	1igh Press	44 We 1774 1875 1977		$\begin{array}{c c} \hline \mathbf{R} \\ \hline \mathbf{R} \\ \hline \mathbf{a} \\ \mathbf{a} \\ \hline \mathbf{a} \\ \mathbf{a} \\$	freq.)	J., oint ater ressure
	RESS							S re fa	SRF tress duction ctor

(A)	Locat	ion: SIMP SH OIA	EVARD	De /	pth / chains	ige: 19-9	D P	ate: 13/5/03 age: (2)
Numbers	Q (ty	pical range) =		Q (mea	an) =		Q (most freq	L)=
DF	-)x()x(\rightarrow)×()	x()	()x()x(
under- ne, or	BLOCK	Very Poo	r Pi	xx 3	Fair	759 5	000(111010) Exc. 777949 56817 84772 666771010 66668977 6-5668977 6-5668977 6-5658780 6-443779 7233246 5233246 52332 723 723	RQD % Core piece ≥ 10 cm
pecify)	SINES	0 10 Earth Four 20 15	20 30 Three 55 65 65 65 12 9	40 50 10 77 10	60 70 10 10 10 10 10 10 10 10 10 1	80 One 2 1	00 100	J Number of joint sets
121-7 = 122-7- × 129-0 138-8 = 138-8- 150-0 = 150-0- 160-9 = 160-9 = 160-9 = 171-0 = 183-3 = 183-3- 195-0	FRICTION	1.0 1.0 20 13 12 1	Plan	er 18772 4 9 9 9 38 5 2 7772 4 56 6 6 5 6 7 7 2 7 7 7 7 7 7 7 2 7 7 7 7 7 7 7 7	Undulating 1310 5 deed 3 3 29 6 6 6 6 2 3 3 5 4 4 4 4 5 5 2 Cost 10 11 11 11 5 2 Cost 11 11 5 2 Cost 12 2 Cost 14 4 5 11 11 5 2 Cost 14 4 5 11 11 5 2 Cost 14 5 14 5 14 14 5 14 5 14 14 5 14 14 14 14 14 14 14 14 14 14 14 14 14	710 710 69 443 3 ed 12 72 69 443 5 6 710 6 9 10 6 10 710 6 9 10 6 10 710 6 9 14 15 7 10 15 15 15 15 15 15 15 15 15 15	Disc. 4 4 9 10 55807 10 55807 10 55560 11 12 11 12 11 11 11 10	J, Joint roughness - least favourable Joint alteration - least favourable
Photos or Sketch 195-0− =9929	ACT-VE S	Exc. Infl	0.2 0.3	h Press 76 56564 563/1 3 0.5 Fault	Viet 05 002 5732 5732 5732 5732 5732 5732 5732 5732 5757 54666 52575 1648 53555 1648 53555 1648 53555 1648 53555 1648 165 1757 1648 165 1757	Pri 27 197 2 2 0 2 0 2 0 2 0 2 0 2 0 2 0 2 0 2 0 2	$\frac{\mathbf{Y}}{\mathbf{D}} = \begin{bmatrix} \mathbf{R} \mathbf{Q} \mathbf{D} \\ \mathbf{J}_{\mathbf{R}} \end{bmatrix} = \begin{bmatrix} \mathbf{U} \\ \mathbf{J}_{\mathbf{R}} \end{bmatrix}$ $\frac{\mathbf{J}_{\mathbf{R}}}{\mathbf{J}_{\mathbf{R}}} = \begin{bmatrix} \mathbf{U} \\ \mathbf{J}_{\mathbf{R}} \end{bmatrix}$ (most freq enoth $\mathbf{I}_{\mathbf{R}} = \begin{bmatrix} \mathbf{U} \\ \mathbf{U} \end{bmatrix}$	Joint Water pressure

(A)	Locat	SH OLA	EVAN	LP.		Depth.	-9	-302	ġ		Page	= 12/5/0 = (3)
umbers	O (tv	pical range) =		-	00	nean):	-			Q (mos	t frea.)	-
r	6	14	141)	L) >(-) 1	_	-	11	1×1
omains,	(M	141	1	11	14		1~1	/	(101	14
re		Very Poor	<u> </u>	Poor	2	-	ar	1	Goo	2.0	Exc	
nnel					-	_	-			-		ROD
ngths					_	_	-		-	10		Core piec
-	в					-	-	4	89.10	101010	10.10	≧ 10 cm
35	L					A 0 10	-	BRA	7.65	3357	7891	
nder-	0 C				181	thi	117	663	3.533	344	444	
e, or ecity)	ĸ	0 10 3	10 3	0 40	5	2.6 2	222	0 0	17.111	90 1	<u>マダダ季</u> 00 100	
GCSYI		Earth Four		Three	52/1	Two	_	One	1	None	1	
	ĩ	Land Loss	12	99200		-12	2	3				
	Z			89889 4	110			-				J.,
	S			7737 7	15	-	-		-			Number
				2556 7	3							joint sets
199-9-	1		1 8 3 4	3333 6	SEC	77	-					
205-1		8.5	666	228/ 3	211	445	it to	u .				
205-1-	-	20 15	12	9	6	4	3	2	1	0.5		
215-7		Filts		Planar		ų	ndulati	39	1	Disc.		
215-7-		2		10	26	-	1010 10.10	12.				1.1
126-2				1.	418 4		1991			-		J,
216-2-				7	1127	-	6677					roughner
357-5	F			10 4	142			28		-		- least
71310	R			444	1114		4445	9910		-		Nivourao
712.0	ċ	66		444 2	1131		(112	1112		-	1	
764-6	Ť	1.0	0.5	1	1.5	1.5	2	3		4		
359.1-		Thick P	100		Then P	115	7.9	eted	19.9	2.4	Phoal	1
270.2	Ň						-	-	1010/04	10.010	-	Joint
270-2-	1						-	-	3525	55555		alteration
181.0				FF		TI	1012	-	1077	4464		- least favourab
281-0-	1.1				- 6		37/7		666.3	3/378		Laterate
242.0			6.5				1124	133.0	1222	113.12	0.75	
hotos	-	20 13 12 10	8 0	5 1Z	0 0	1	4	3	4	1	0.75	-
		Exc. Inno	WS.	nyter	11	70	797 KB	ANRAS	LIY	BOD	-	
or				-	-	1	77794	7999	-	1,		يد ا
sketch	A				-	1	採購	144			-	Joint
2-0-	Ç			-	_	1	4445	10	-	1 .	-	pressure
4= P.50	i			-	159.9	1	11111		4/11.11		-	
	Y			7	789	-	21227		4440			
	-	0.05 0.1	0.2	0.33	0.5		0.66	·	1	a (mo	er med)	
	S	Squeeze	1,9	Mon	,F	aults	IF.	Street	s/Stren	gihi 1	2	
	R									1010	#30	SRF
	E								-	244	11	Stress
	S						H		-	2747	10	reduction
	~								-	1202	44	Tuctor
										22.11	1112	
		20 15 10 5	20 15	10 5	10 7	5 5 2	5 100	50 20	5 2	0.5 1	2.5	

IS-	Locat	K2H	DIA	P.		302-4-	- 398-6			Page	× (4)
lumbers	Q (ty	pical range) =	=		Q (m	ean) =			Q (most	freq.) =	-
omains	(-)x()x()	6)×()×(\rightarrow	()	×()x(
ore oxes, unnel engths under- ne, or percifu)	BLOCK		20 3	Poor	66 1	Fair 5	12.61	Go		Exc.	RQD % Core piecr ≩ 10 cm
= 3024- 313-4 = 313-4	SIZES	20 15	4 6 6 6 2 12	Three 111 111 111 111 111 111 111	1944 1944 1948 1948 1948 1948 1948 1948	7 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	2	1	None		J _n Number o joint sets
325-1 335-6 = 335-6- 348-5 = 348-5 = 348-7 = 348-7	FRICTION	1.0 20 13 12 1	0.5 Fills	Planer 4 1 2 5 12 1 1 1 1 1 1 1 1 1 1 1 1 1	5944 5944 555 555 555 555 555 555	Undular 9 9 9 9 9 9 9 9 9 9 9 9 9	ting 1 1 1 1 1 1 1 1 1 1 1 1 1		4 milled 4 4 4 4 4 4 4 4 4 4 4 4 4	Heat	J, Joint roughnesi - keast favourable Ja Joint afteration - keast favourable
Or Sketch	ACT-VE STRESS	0.05 0.1	0.2	High Pri	99 6 7 0.5 Fa	C2 Wet 14443 77,75 6752 72,755 14444 3555 14444 3555 14444 1355 111111 0.060 atts	Sire	289 289 265 235 235 235 235 235 235 235 235 235 23		(freq.)	Juint Water pressure SRF Stress reduction factor

(K)	Locat	KSH 0	IA	· ·		398.8	- 50	0.1		Page	5
umbers	Q (by	pical range) =			Q (m	ean) =			Q (mos	st freq.)	-
r omains.	()×(-)×(-		(-)x()×(-)	()x()×(
nre xxes, nnel ngths	1	Very Poor	2	2 2	Б	s 6	1	Go	27F	Exc. 270	RQD 1
nder- e, or ecily)	BLOCK	3331	÷1 20 3	33 2 3 0 40	6 4.4 334 3 50	457 4	715 15 15 12 70	13 1 7 18 1 7 6 5 80	7 227 7 227 7 25 25 7 25 27 7 25 27 7 23/// 90 1	10 10 11 19 17 19 17 7.58 23566 11 272 00 100	<u>≧ 10 cm</u>
	SIZES	Earth Four	75	Three 44491 11588 28 7277 23666	1	Two	One		None		J_ Number joint sets
398-8- 4/0-4 4/0-4-		20 15	976 6543 33211 12	9 9	7979 22809 4-554 54 6	H He Ju 566 HB 233 [21] 4 3	2	1	0.5		
421-1 421-1- 431-3 431-3- 441-7		Fills		Planar 10 17 17 10 17 10 10 17 10 10	10 20 99 00 84499 8888 6679	Undui (4) 200 日中 (7日 (4) (4) (4) (4) (4) (4) (4) (4) (4) (4)	100 1/2		Disc		J, Joint roughne
4417- 4622 4572-	FRICTI	54 1.0 Thick F	0.5	72874	455 1.54 1.5 Thin Fil	45 199 393 393 115 2 5	56 44 56 40 3455 2 7222 3 Couled		4_	Heat	fevourat
462-8- 473-5 473-5 484-2 484-2 484-2	ON		5.9		44		4191 187 727 7334 4 3341	15 15 15 15 15 15 15 15 15 15 15 15 15 1	9494011 1000 99 740 90 740 90 85555 255555 231404 93351 3711172		J _a Joint alteratio - inast favourat
or ketch 4448- 500-1	ACT->	20 13 12 10	8 6	5 12 High Pre 94 94 94 94 94 94 94 94 94 94 94 94 94	8 6 164 164 164 164 164 164 164 16	4 4 W	3	2 Dry 2.5 18 19 5.7 2.5 18 5.7 2.5 18 5.7 2.5 18 5.7 2.5 18 18 5.7 2.5 18 18 5.7 2.5 18 18 18 18 18 19 19 19 19 19 19 19 19 19 19 19 19 19	RQD	0.75	J., Joint water pressure
3,4,5 16.5mpl)	STRESS	0.05 0.1 Squeeze	0.2 Sa	0.33	0.5 Fa		Sin	1 HSS/Clime	1 (mo	er med')	SRF Stress reductio factor
		20 15 10 5	20 15	10 5	10 7.5	73 73 5 2.5 1	00 50 20	5 2	0.5 1	2.5	

<	Locati	ion: SIN KS	HOIA	RP		Depth	/ chai	nage: - 60	0.6		Date: Page	13/5/03
Numbers	Q (ty	pical rang	e) =] Q	(mean)	=			Q (mos	st freq.) =	=
for	(-	-)x(-)	x($\rightarrow ($)x	()x(\rightarrow	()x()x(
core boxes, tunnel lengths (under- line, or	81005	Very	Poor		'oor ⇒€¶₽5	10 10 589	Fair 5	7.9.0	G00	974 974 00 96 45 56 444 3444 312	Exc. 18 (8) 997.16 18/9 8/9 8/9 8/9 8/9 8/9 8/9 8/9 8/9 8/9	RQD % Core piece: ≧ 10 cm
1=5001- 505-// 2=505-0-	5-NES	0 10	20 Four	30 The 10 1700 700 1000 700 1000 700 1000 700 1000 700 1000 700 1000 700 1000 700 1000000	40 60 75 75 75 75 75 75 75 75 75 75	50 6 Two 10 10 10 10 10 10 10 10 10 10	20 70 77 88 85 17	70 (One		90 1	00 100	J _n Number of joint sets
515-9 526-3 526-3 4=526-3 536-5 536-5 536-5 546-8	FRI	Fills		Plan	MM - 19 (2) -		9 10 9 10 9 10 9 10 9 10 9 10 9 10 9 10	ng 12 13 9 9 9 9		Disc.		J, Joint roughness - least favourable
≥546 € 5578 5679 ≥5679 ≥5679 ≥5787 ≥5787 ≥5885 ≥5885	CH-OZ		hick Fills		1.5 Thin 13 8	1.5 File	2 04 1010 99 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10 1	3 aled 10 10 10 10 10 10 10 10 10 10 10 10 10	Un		Heal	J _a Joint alteration - least favourable
Photos or Sketch =5.81-3- 600-6 P (3.5et) 314 P (4.set)	ACTIVE ST	0.05	0.1 0	Hi 12 0:3 Swell	29 Press	Faults	Wet 13 10 14 10 10 2 2 2 4 9 5 64 4 6 6 40 3 3 5 5 3 1 1 1 1 (11 1 2) 0.06	Sine	Dry 25561 25561 1121 1 1555	ROD 	st freq.)	Ju Joint water pressure
541 (1+112) 563 (4+1) 563 (4+1)	RESS									14 1772 1472 1472 1446 1905 1446 1905 1446 1905 1446 1005		SRF Stress reduction factor

F25:541-570

(A)	Locat	KSH (EVARP		Depth / chu 600-6	inage: - 697	s		Date: Page	13/5/03
umbers	Q (ty	pical range) =			nean) =			Q (mos	t freq.) =	-
Nr.	(-)x()×(\rightarrow)x()x(\rightarrow	()x()×(
ore oxes, unnel engths under- ne, or	BLOCK	Very Poor	2.333.85	200r	8 19 4-9 13 4-112	1 555 + 11234	G00 14 9 3 5 7 7 2 3 4 2 4 3		Enc.	RQD % Core pieco ≥ 10 cm
= 600-6- 501-4 = 611-4-	SIZES	0 10 Earth Four 20 15	20 30 Thu 77 2 86 497 37 66 56 956 957 80 1957 80	40 30 88 75 67 67 10 11 12 31 14 36 8	00 Tero 2 5 u q q y 7 5 u q y 10 5 u q y 10 10 10 10 10 10 10 10 10 10	2	1	None	100	J _n Number o joint sots
622-1 =622-1 632-5 643-5 643-4 653-7 =653-7 =653-7	FRIC	Fills	Pla	107 25 129 75589 75556 65666 5566 45555 564433 38 3153 1077112	Unduk 23 10 % 24 % 24 % 25 % 24 % 24 % 24 % 24 % 24 % 24 % 24 % 24	110g		Disc		J, Joint roughnes - least favourabl
663.6 = 663.6 694.2 = 694.2 684.2 684.2 684.2 684.2 684.2	H-OZ	1.0 Thick I 20 13 12 10		1.5 Thin F		3 2001ed 9971 9777 9777 9777 9777 9777 9777 977	Un 1562 1562 1105 1105 1105 1105 1105 1105 1105 110	4 110 110 110 110 110 10 10 10 10 10 10 1	Heal	Joint alteration - least favourabl
or Sketch 614-1- =6995	ACT-VE S	Exc. Infic	0,2 0.	ip ip 4 E 4 74 i ip 3 3 2 11 33 0.5	Viet 10 10 10 10 10 10 10 10 10 10	Street	904 2.44 5.666 11-14 1 1553ren	$\frac{RQD}{J_{a}} = \frac{1}{J_{a}} = 0$ (more standing of the stand	et freq.)	J _w Joint water pressure
	TRESS	20.45.40.5	20 45 10	5 40 7	1	0.50.20			28	SRF Stress reduction factor

(A)	Locat	ion: SIMPE KSH 0	VARP	la,		Depth 6	/ chair 17 · ⊆	nage; 	92.8		Page	14/5/03
umbers	Q (by	pical range) = [00	nean)	=	and the second s		Q (mos	t freq.) =	
r	(-)x()x()	6	-)x()x(-	()x()x(
omains, ore oxes, innel ingths	BL	Very Poor		Poor		3	an .		Goo	0 10101049 1410 1410 1410 1410 1410 1410	Exc 7 4480 4469779 56660 555500 55550 55550 555500 555500 555500 555500 555500 555500 555500 555500 555500 55000 55000000	RQD % Core pieco
inder- ne, or becify)	OCK S-NES	0 10 2 Earth Four	0 30) 40 Three	45	2 34 0 60 Two 0	23 17 23 17 24 18 10 77 18 10 77 18 10 77 28	0 8 0 8 0 8 0 8 0 8 0 8 0 8 0 8 0 8 0 8	422 0	543 332 1//2 90 10 None	3934 2231 11224 00 100	J _a Number o joint sets
= 699.5- 7028 =704-8 =104-8		20 15	44 37 12	444 233 212 9	10 7 446 33 1 6	1444 7444 7444 7444 7444 1112 4	455 455 334 123 3	2	ष् हेव्यू हेव्यू 1	0.5		
7/49- 9/49- 735-1 735-1 7459 =7459	FRIC	Fills	0.5	Planar KE GLE33	10 10 10 10 10 10 10 10 10 10 10 10 10 1	15	0000000 100000 100000 100000 10000 10000 10000 10000 10000 1000	17 17 17 17 17 17 17 17 17 17 17 17 17 1		ORSC		J _e Joint roughnes - least favourabl
2562 7562 7667 7667 7767 7777 7777 7777	0N	20 13 12 10	15	5 12	Thin F		200 2 10 10 1 12 4 9 9 4	10 13 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10 10	Un	1000 001 11000 000 11000 000 1100000 110000000000	Heal	J _a Joint alteration - inast favourabi
or Sketch 768-1- 798-8	ACT-VE	Exc. inflo	12	High P	8.97 1444 1331		Wei 11.14 11.1	10	- Dry - 9 10 0 10 979 404 977 404 977 788 977 788 965 67 565 66 555 55 64 0 55 7 1 1 8 1 1 1 1 1		st freq.)	J _w Joint water pressure
	ST R E S S	Squeeze	9.2 		F	aults	0.66	Stre	Stron	ah a 1 Pau 12 Sav Sav Sav Sav Sav Sav Sav Sav Sav Sav		SRF Stress reduction factor
		20 15 10 5	20 15	10 5	10 7	5 5 2	5 100	50 20	5 2	0.5 1	2.5	

(A)	Locat	ion: 51	MPE	VAR	P		Depth	/ chair 79	age:	901-	2	Page	14/5/03
umbers	O /by	nical rar	(en	GIA	-	100	neani	=	0 0		O (mos	t freq) =	
r	6)v/	de1 -1	14	-)vi	_	111	_	6	17(111
omains, ore oxes, innel ngths		Ve	ny Poor		Poo			-air		Goo	25 IN	Exc. 19171 - 1-1 14506 - 11 7 2 8 86 7 7 7 7 7 7 2 8 86 7 7 7 7 7 7 1 6 6 6 7 7 7 7 7 7 1 6 6 6 7 7 7 7 7 7 1 6 6 6 7 7 7 7 7 7 1 6 7 7 1 6 7 7 7 8 8 7 7 7 8 7 7 7 8 8 7 7 7 8 7 7 7 8 8 7 7 7 8 7 7 7 8 8 7 7 7 7 7 8 8 7 7 7 7 7 7 8 7 7 7 7 7 7 8 7 7 7 7 7 7 7 7	RQD % Core piece
nder- ie, or iecify)	BLOCK S	0 1 Earth	0 2 Four	20	30 4 Three	10 5	0 6 Two	0 7	3 70 (One	10 3 4-9-56 10 1	7774 C556 4445 12133 90 1 None	22576 23345 2255 171111 171111 171111 171111 171111 171111 1711111 1711111 17111111	
778-2- 201-1 201-1	ZES			2		- 3	494 794 56555 4 7 11	14 1919191 1 2 2 2 1 2 2 2 2	10 2021 2021 2021 2022 2022 2022 2022 2	13			J _n Number joint sets
8/1-7- 8/1-7- 830-1 850-1- 840-8 840-8- 851-9	FR-	Filh	15	12	Planar Planar 7276 6656 5443 2951	0 449 55555 55555 4406 55555 4406 55555		3 Incluiation Incluiation Incluiation Incluint I			Disc.		J, Joint rougtnes - least favourabl
851-9- 863-5- 873-6 873-6 873-6 884-7 884-7 884-7 884-7 884-7	UT-ON	1.0	Thick F	0.5		1.5 Thin F	15	2 0	3 ated 399 7988 7988 7988 7988 7988 7988 7988	S Un	4 Head 1=10 Exercises Exercises 2777280 Exercises Exerci	Heal 1964	Joint atteration - least favourab
or iketch Resch Pol+2	ACT-VE	E	xc. inflo	Wis	High	Pmss		219 219 275 275 275 2455 2455 2455 2455 2455 24		Pry 777777 565555 565555 775555 7555555	RQD	eno	J _w Joint water pressure
	STRESS	0.05 Squ	0.1	0.2	0.33 Swell	0.5	aults	0.66	Stre	t ss/Siren	010 45 0210 45 0211 45 0211 15 0211 15 0000000000000000000000000000000000		SRF Stress reduction factor
		20.15	10 5	20 1	5 10 5	10.7	5.6.2	5 100	50 30	5 2	0.6 1	2.5	

dias-	Locat	ion: S	IMPE	VAL	2 P		Dept	/ chai	nage:			Date:	14/5/00
~		K	3110	MM		11.	-16	11 6	100			Fage	0
lumbers	Q (ty	pical ran	(ge) = [_	_	10(mean)	=			Q (mos	st freq.) =	L
iomains	(-)x(-)×(-) (-	—)x)×(\rightarrow	()×()×(
ore oxes, unnel ongths	в	Ve	ry Poor	1	Poor			Fair		Goo	79)	19 Exc. 19 24492342 (4) 24 (24 (4) 24 (24 (4) 24 (24 (4) 25 (5) 24 (25 (5) 24 (25 (5) 24 (25 (5)	90 RQD % Core piece ≩ 10 cm
under- ne, or pecify)	LOCK S-NES	o 1 Earth	0 2 Four	D	30 4 Three	0 5	0 (Two	2	70 0 000	77 9 1010 19 10 19 10 19 10 19 10 19 10 10 10 10 10 10 10 10 10 10 10 10 10 1	# 559 4-45 3-22 90 1 None	127334 33633 32272 001111 500	J _n Number o joint sets
= 901-2 * 906-1 = 906-1-		20	15	12	9	62	755	7766 64655 4513 22.2.3 3	12024 12024 13133 1272 1121	55 45 370 270 1	9 9 8 5 0.5		
= 916.7- 927.8 = 927.8 938.8 = 938.8 949.9	FRIC	Fills			Planar	23	56	171000000 171000000 10200000 10200000 10200000 10200000 1020000 1020000 10200000 10200000 102000000 1020000000 10200000000		+	10 11 11 11 11 11 11 11 11 11 11 11 11 1		J, Joint roughness - least favourable
960-9 972-0 972-0 972-0 972-0 972-0 972-0 972-0 972-0 972-0 972-0	TION	1.0	12 10	0.5		1.5 This 2 8	1.5 785 1- 1- 1- 1- 1- 1- 1- 1- 1- 1- 1- 1- 1-	11 2 4 Co	3 ented 41.0 67.0 67.0 80 77.35 3		1 4 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	Heal	J _e Joint alteration - least favourable
or Sketch	ACT	6	sc. inflov	6	High	Press		10 10 10010	72	17 Dry 1777 00 601 20 601 20 60 5 72 61 5 20 7 3 55 5	ROD		Joint water pressure
END	VE STRESS	0.05 Squ	0.1	0.2	0.33 Swoll	0.5	aults	2444	Stree	71111 13151 101 1055/ref	din (me	st freq.)	SRF Stress reduction factor

(B)	Locatio KSH	OLA m	MPEVA	RP	- Dep	th / cha 3 to /	inage: 003-0	m		Date: Page	14/5/03
Numbers for	Q (typ	ical range)x()=[)x()		Q (mear ()	n) = [×()x(Q (mos	st freq.) = -)x()x(
core boxes, tunnel lengths (under- line, or	THRAPPART	Vory F	2007	Poor 1 1 2 3 3 3 4 4 1 2 3 3 4 4 1 2 3 3 4 4 1 2 3 3 4 4 1 2 3 3 4 4 1 1 2 3 3 4 4 1 1 1 2 3 3 3 4 4 1 1 1 1 1 1 1 1 1 1 1 1 1	-ME-103-30M	Fair 5 / 6 5 6 5 6 5 6 5 6 5 6 5 6 5 6 5 6 5		Gox 12 9 16 7 10 2 15 5 6 2 2	1 35 477 333 477 333 477 333 477 247 247 25 25 25 25 25 25 25 25 25 25 25 25 25	Exc. 1238 388 227 7/6 229 7/6 229 7/6 229 7/6 229 221 58 209 7/6 229 7/6 200 7/6 7/2 200 7/6 7/2 200 7/6 7/6 7/2 200 7/6 7/2 200 7/6 7/2 200 7/6 7/2 200 7/6 7/7 200 7/6 7/7 200 7/6 7/7 7/7 7/7 7/7 7/7 7/7 7/7 7/7 7/7	RQD % Core piece ≧ 10 cm
1=1.3 - 1 00-8 2=100.8 -	S-NES	Earth F	20 3 Dur 5 6 5 5 13 13 13 13 14 14 1 14 15 12	Three 2.5 2.1 17 2.1 17 2.1 42.5 2.2 44.5 2.4 44.5 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.4 10 1.5 2.2 9 6	1 27 3 9 23 3 4 2 3 1 4 2 4 5 1 3 4	7-4 14 man (* 14 14 14 14 14 14 14 14 14 14 14 14 14	0mm 1 21 2.7 4.4 2	4 222 1	None 6 0.5		J _a Number of joint sets
199-9 302.9 4=302.9- 398.8 5=398.8- 500.1-	La nan antique	Fills 		Planar 4 14 333 10 399 14	43	Undulat ALL Press Birds Cole	ing 22/32/21/2 1/2/11/05/33/8/6		Disc.		J, Joint roughness - least favourable
600-6 1= 600.6- 699.5- 798.8 ≈ 7988- 7988-	- " A SWAP RF 2	20 13 12	0.5 0.5 0.5 0.5 1 1 10 8 6		nin Fills nin Fills 1 1 1 1 1 1 1 1 1 1 1 1 1		001ed 23 17 17 17 17 17 17 17 17 17 17	Dopanin Syder and N	111100 m 0.4 % 0 1 1 0 % 0 4 1 0 % 0	Heal	J _a Joint alteration - lenst favourable
or Sketch 0= 90/2- 1#03.0 END		Exc.	inflows	High Pres 4 12 12 12 5 2 2 5 3 9 1 0 1 5 3 0 1 5 1 5 1 0 1 5 3 0 1 5 1 5 1 0 1 1 5 1 1 2 1 2 5 5 1 2 5 5 1 2 5 5 1 2 5 5 1 2 5 5 1 2 5 5 1 2 5 5 5 1 2 5 5 5 1 2 5 5 5 5	5	Vitt 61 61 62 62 62 62 62 62 62 62 62 62 62 62 62		Dry 24.0 24.0 153 257 193 552 179 5352	RQD	st freq.)	J. Joint water pressure
	~ H - H - H - H - H - H - H - H - H - H	Sourcez		0.33 0	Faults	2 44	Stre	1 ss/3brei	91 65 90 91 91 92 95 95 95 95 95	22010	SRF Stress reduction factor

Q-method of rock classification

Q-logging ratings for RQD, J_n , J_r , J_a , J_w and SRF (Barton, 2002)

$$Q = \frac{RQD}{J_n} \times \frac{J_r}{J_a} \times \frac{J_w}{SRF}$$

RQD is the % of competent drill-core sticks > 100 mm in length [1] in a selected domain

 J_n = the rating for the number of joint sets (9 for 3 sets, 4 for 2 sets etc.) in the same domain

 J_r = the rating for the roughness of the least favourable of these joint sets or filled discontinuities

 J_a = the rating for the degree of alteration or clay filling of the least favourable joint set or filled discontinuity

 J_w = the rating for the water inflow and pressure effects, which may cause outwash of discontinuity infillings

SRF = the rating for faulting, for strength/stress ratios in hard massive rocks, for squeezing or for swelling

RQD / J_n = relative block size (useful for distinguishing massive, rock-burst-prone rock)

 J_r / J_a = relative frictional strength (of the least favourable joint set or filled discontinuity)

 J_w / **SRF** = relative effects of water, faulting, strength/stress ratio, squeezing or swelling (an 'active stress' term)

An alternative combination of these three quotients in two groups only, has been found to give fundamental properties for describing the shear strength of rock masses – something close to *the product of 'c' and 'tan \varphi'*. By implication Q (and in particular Q_c) have units resembling *MPa*.

Footnotes below the tables that follow, also give advice for *site characterization* ratings for the case of Jw and SRF, which *must* not be set to 1.0 and 1.0, as some authors have suggested. This destroys the intended multi-purposes of the Q-system, which has an entirely different structure compared to RMR.

1. F	Rock Quality Designation	RQD (%)
А	Very poor	0-25
В	Poor	25-50
С	Fair	50-75
D	Good	75-90
Е	Excellent	90-100

Notes:i) Where RQD is reported or measured as \leq 10 (including 0), a nominal value of 10 is used to evaluate Q.

ii) RQD intervals of 5, *i.e.*, 100, 95, 90, etc., are sufficiently accurate.

2. J	2. Joint set number		
Α	Massive, no or few joints	0.5-1	
В	One joint set	2	
С	One joint set plus random joints	3	
D	Two joint sets	4	
Е	Two joint sets plus random joints	6	
F	Three joint sets	9	
G	Three joint sets plus random joints	12	
Н	Four or more joint sets, random, heavily jointed, 'sugar- cube', etc.	15	
J	Crushed rock, earthlike	20	

Notes: i) For tunnel intersections, use $(3.0 \times J_n)$.

ii) For portals use $(2.0 \times J_n)$.

3. J	3. Joint roughness number						
a) I	a) Rock-wall contact, and b) Rock-wall contact before 10 cm shear						
Α	Discontinuous joints	4					
В	Rough or irregular, undulating	3					
С	Smooth, undulating	2					
D	Slickensided, undulating	1.5					
Е	Rough or irregular, planar	1.5					
F	Smooth, planar	1.0					
G	Slickensided, planar	0.5					

Notes: i) Descriptions refer to small-scale features and intermediate scale features, in that order.

b) No rock-wall contact when sheared					
Н	Zone containing clay minerals thick enough to prevent rock- wall contact.	1.0			
J	Sandy, gravely or crushed zone thick enough to prevent rock-wall contact	1.0			

Notes:ii) Add 1.0 if the mean spacing of the relevant joint set is greater than 3 m.

- iii) $J_r = 0.5$ can be used for planar, slickensided joints having lineations, provided the lineations are oriented for minimum strength.
- *iv)* J_r **and** J_a classification is applied to the joint set or discontinuity that is least favourable for stability both from the point of view of orientation and shear resistance, τ (where $\tau \approx \sigma_n \tan^{-1} (J_r/J_a)$).

4. J	oint alteration number	<mark>ф</mark> г approx.	Ja					
a) F	a) Rock-wall contact (no mineral fillings, only coatings)							
А	Tightly healed, hard, non-softening, impermeable filling, i.e., quartz or epidote.		0.75					
В	Unaltered joint walls, surface staining only.	25-35°	1.0					
С	Slightly altered joint walls. Non-softening mineral coatings, sandy particles, clay-free disintegrated rock, etc.	25-30°	2.0					
D	Silty- or sandy-clay coatings, small clay fraction (non- softening).	20-25°	3.0					
Ш	Softening or low friction clay mineral coatings, i.e., kaolinite or mica. Also chlorite, talc, gypsum, graphite, etc., and small quantities of swelling clays.	8-16°	4.0					
b) Rock-wall contact before 10 cm shear (thin mineral fillings)								
F	Sandy particles, clay-free disintegrated rock, etc.	25-30°	4.0					
G	Strongly over-consolidated non-softening clay mineral fillings (continuous, but < 5 mm thickness).	16-24°	6.0					
Н	Medium or low over-consolidation, softening, clay mineral fillings (continuous, but < 5 mm thickness).	12-16°	8.0					
J	Swelling-clay fillings, i.e., montmorillonite (continuous, but < 5 mm thickness). Value of J _a depends on per cent of swelling clay-size particles, and access to water, etc.	6-12°	8-12					
c) N	lo rock-wall contact when sheared (thick mineral filling	s)						
KL M	Zones or bands of disintegrated or crushed rock and clay (see G, H, J for description of clay condition).	6-24°	6, 8, or 8-12					
Ν	Zones or bands of silty- or sandy-clay, small clay fraction (non-softening).		5.0					
OP R	Thick, continuous zones or bands of clay (see G, H, J for description of clay condition).	6-24°	10, 13, or 13-20					

	5. Joint water reduction factor	approx. water pres. (kg/cm ²)	J_w
Α	Dry excavations or minor inflow, i.e., < 5 l/min locally.	< 1	1.0
В	Medium inflow or pressure, occasional outwash of joint fillings.	1-2.5	0.66
С	Large inflow or high pressure in competent rock with unfilled joints.	2.5-10	0.5
D	Large inflow or high pressure, considerable outwash of joint fillings.	2.5-10	0.33
Е	Exceptionally high inflow or water pressure at blasting, decaying with time.	> 10	0.2-0.1
F	Exceptionally high inflow or water pressure continuing without noticeable decay.	> 10	0.1-0.05

Notes: i) Factors C to F are crude estimates. Increase J_w if drainage measures are installed.

ii) Special problems caused by ice formation are not considered.

iii) For general characterization of rock masses distant from excavation influences, the use of Jw = 1.0, 0.66, 0.5, 0.33 etc. as depth increases from say 0-5m, 5-25m, 25-250m to >250m is recommended, assuming that RQD /Jn is low enough (e.g. 0.5-25) for good hydraulic connectivity. This will help to adjust Q for some of the effective stress and water softening effects, in combination with appropriate characterization values of SRF. Correlations with depth-dependent static deformation modulus and seismic velocity will then follow the practice used when these were developed.

6. S	Stress Reduction Factor	SRF				
a) Weakness zones intersecting excavation, which may cause loosening of rock in when tunnel is excavated						
А	Multiple occurrences of weakness zones containing clay or chemically disintegrated rock, very loose surrounding rock (any depth).	10				
В	Single weakness zones containing clay or chemically disintegrated rock (depth of excavation \leq 50 m).	5				
С	Single weakness zones containing clay or chemically disintegrated rock (depth of excavation > 50 m).	2.5				
D	Multiple shear zones in competent rock (clay-free), loose surrounding rock (any depth).	7.5				
Е	Single shear zones in competent rock (clay-free), (depth of excavation \leq 50 m).	5.0				
F	Single shear zones in competent rock (clay-free), (depth of excavation > 50 m).	2.5				
G	Loose, open joints, heavily jointed or 'sugar cube', etc. (any depth)	5.0				

Notes: i) Reduce these values of SRF by 25-50% if the relevant shear zones only influence but do not intersect the excavation. This will also be relevant for **characterization**.

b) C	Competent rock, rock stress problems	σ_c/σ_1	$\sigma_{\theta}/\sigma_{c}$	SRF
Н	Low stress, near surface, open joints.	> 200	< 0.01	2.5
J	Medium stress, favourable stress condition.	200-10	0.01-0.3	1
к	High stress, very tight structure. Usually favourable to stability, may be unfavourable for wall stability.	10-5	0.3-0.4	0.5-2
L	Moderate slabbing after > 1 hour in massive rock.	5-3	0.5-0.65	5-50
М	Slabbing and rock burst after a few minutes in massive rock.	3-2	0.65-1	50-200
Ν	Heavy rock burst (strain-burst) and immediate dynamic deformations in massive rock.	< 2	> 1	200-400

Notes: ii) For strongly anisotropic virgin stress field (if measured): When $5 \le \sigma_1 / \sigma_3 \le 10$, reduce σ_c to 0.75 σ_c . When $\sigma_1 / \sigma_3 > 10$, reduce σ_c to 0.5 σ_c , where σ_c = unconfined compression strength, σ_1 and σ_3 are the major and minor principal stresses, and σ_0 = maximum tangential stress (estimated from elastic theory).

- *iii)* Few case records available where depth of crown below surface is less than span width. Suggest an SRF increase from 2.5 to 5 for such cases (see H).
- *iv)* Cases L, M, and N are usually most relevant for support design of deep tunnel excavations in hard massive rock masses, with RQD /Jn ratios from about 50 to 200.
- v) For general characterization of rock masses distant from excavation influences, the use of SRF = 5, 2.5, 1.0, and 0.5 is recommended as depth increases from say 0-5m, 5-25m, 25-250m to >250m. This will help to adjust Q for some of the effective stress effects, in combination with appropriate characterization values of Jw. Correlations with depth dependent static deformation modulus and seismic velocity will then follow the practice used when these were developed.

c) S in	queezing rock: plastic flow of incompetent rock under the fluence of high rock pressure	$\sigma_{\theta}/\sigma_{c}$	SRF
0	Mild squeezing rock pressure	1-5	5-10
Ρ	Heavy squeezing rock pressure	> 5	10-20
	1/3		

Notes: vi) Cases of squeezing rock may occur for depth $H > 350 Q^{1/3}$ according to Singh 1993 [34]. Rock mass compression strength can be estimated from SIGMA_{cm} $\approx 5 \gamma Q_c^{-1/3}$ (MPa) where $\gamma =$ rock density in t /m³, and $Q_c=Qx\sigma_c/100$, Barton, 2000 [29].

d) Swelling rock: chemical swelling activity depending on presence of water		SRF
R	Mild swelling rock pressure	5-10
S	Heavy swelling rock pressure	10-15