

## In-situ stress measurements by hydraulic fracturing method at Gotvand Dam site, Iran

Reza ZIAIE MOAYED<sup>1,\*</sup>, Ehsan IZADI<sup>2</sup>, Mohsen FAZLAVI<sup>3</sup>

<sup>1</sup> *Civil Engineering Department, Imam Khomeini International University, Qazvin-IRAN*  
*e-mail: R\_ziaie@ikiu.ac.ir*

<sup>2</sup> *Civil Engineering Department, Imam Khomeini International University, Qazvin-IRAN*

<sup>3</sup> *Earthquake Research Center, Tehran - IRAN*

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### Abstract

A set of hydraulic fracturing tests were conducted for determining in situ stresses in 10 inclined and vertical boreholes ranging in depth from 30 to 100 m at locations of hydropower plant tunnels of the Gotvand Dam site in the southwest of Iran. The rock in the studied area is formed by the sequence of sandstone, clay stone, and mudstone layers of the Aghajari (AJ) formation, and by thick conglomerate layers of the Bakhtiari (BK) formation. Note that the AJ formation is located underneath the BK formation. The in situ tests were carried out by using conventional systems with special high pressure rods and straddle packer tools. The tests were conducted both in fractured and intact rocks. The measurements yielded an orientation of the maximum horizontal stress of  $N 30^\circ \pm 5^\circ$  (NE-SW), which is generally the same as directions indicated by focal mechanisms solutions from earthquakes and fault conditions in the location of the area studied.

**Key Words:** Gotvand Dam, in situ stress, hydraulic fracturing test

### 1. Introduction

Knowledge of field stresses is very important in many problems dealing with rocks in civil, mining, and petroleum engineering. The modern rock mechanics engineer has to be well acquainted with the basis of rock stresses and rock stress measurements. The need for understanding in situ stresses in rocks has been recognized by geologists and engineers for a long time and many methods have been proposed to measure these stresses since the early 20th century.

Hydraulic injection stress measurements for determining in situ underground stresses are used in different areas such as dam and tunnel construction, earthquake prediction, and oil and gas source designing all over the world (Haimson and Fairhurst, 1970; Cornet and Julien, 1989; Ziaie Moayed et al., 2008; Jeffrey et al., 2010).

The appeal of hydraulic fracturing test (HFT) for in situ stress measurements comes from its relatively simple operation and easy interpretation. This technique was first introduced by Clark in 1949. Then it was developed by Hubbert and Willis (1957), Scheidegger (1962), Kehle (1964), Fairhurst (1964), Haimson

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\*Corresponding author

and Fairhurst (1967 and 1970), Fairhurst (1974), and some others. This technique consists of creating a hydraulic fracture inside a borehole by injecting a fluid inside the sealed interval and determining the mechanical characteristics of rock such as shut-in pressure of fractures and tensile strength of rock.

Different methods of fracturing such as hydraulic test on intact or preexisting fractures, sleeve fracturing, and hydro jacking are designed to study the behavior of principal stresses by means of hydraulic concepts (Baumgärtner and Rummel, 1989).

Several studies have been carried out on hydraulic fracturing methods in Iran. Ajalloeian et al. (2011) reported the results and evaluation of hydraulic jacking and hydraulic fracturing tests conducted on the Aghajari (AJ) formation in different depths. Shadizadeh et al. (2009) performed hydraulic fracturing tests on a Bangestan reservoir site, Ahwaz. They performed an extreme test program to find the best layer to perform hydraulic fracturing tests. They stated that the Ilam formation could be a good candidate for this.

The aim of this study was to analyze and assess the results of hydraulic fracturing tests at the Gotvand Dam site, Iran, to determine the magnitude and orientation of the principal stresses.

## 2. General geology of the region

The Gotvand Dam site is located on the AJ and Bakhtiari (BK) formations as shown in Figure 1. AJ is mainly formed by the sequence of brownish to grayish sandstones with clay stone inter-bedded layers and siltstones. The BK formation is generally formed by thick conglomerate layers with lime and silica cement. These thick layers are generally separated by sandstone and claystone interbedded layers. The rock in the area studied is mainly formed by the sequence of AJ layers, especially the mudstone and siltstone layers. Furthermore, the fault is perpendicular to the dam axes (Ziaie Moayed et al., 2008).

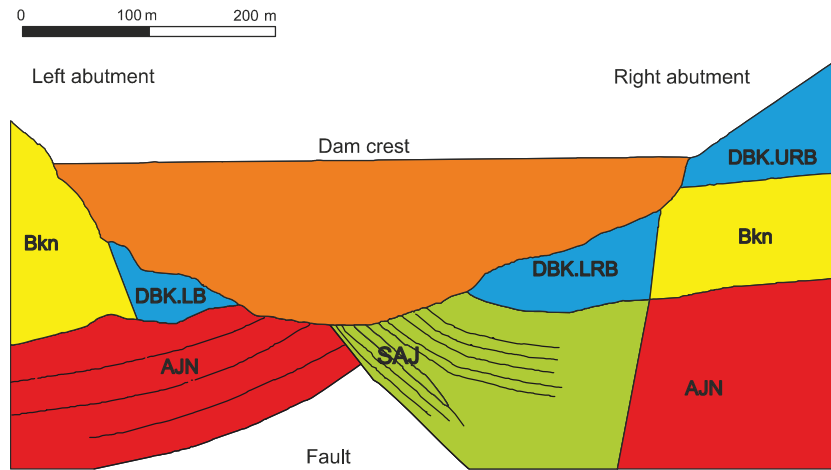
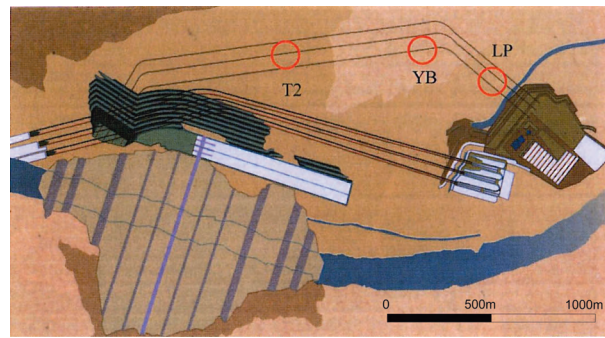


Figure 1. Geological profile in dam axis.

## 3. Drilling and sampling

Borehole drillings and hydraulic fracturing tests were carried out at 3 separate stations of the hydropower plant tunnels of the Gotvand Dam site. The locations of these stations (marked T2, YB, and LP) are shown in Figure 2.



**Figure 2.** Locations of hydropower plant tunnels of the Gotvand Dam site and test stations of HFT.

Ten boreholes ranging in depth from 30 to 100 m were drilled to explore the features of underground layers and to conduct hydraulic fracturing tests. More information related to the boreholes is given in Table 1. A drilling machine (SD300) capable of drilling vertical and horizontal boreholes was used (Figure 3). The applied drilling method was continuous coring with a diameter of 101 mm.

**Figure 1.** Depth and orientation of the boreholes in Gotvand Dam for HFT.

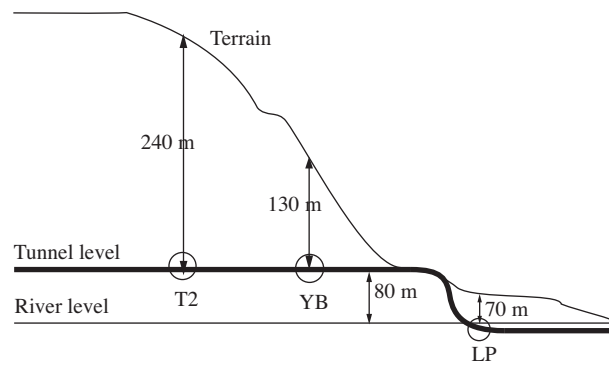
No.	Station	Borehole	DIP	Plunge Direction	Borehole length (m)
1	LP	LPHF1	90	—	60
2		LPHF2	0	—	60
3		LPHF3	90	—	55
4	YB	YBHF1	90	—	50
5		T2HF	90	—	100



**Figure 3.** Drilling machine (SD300) while drilling horizontal borehole.

The YB and T2 stations' elevation is equal to the tunnel level and the LP station has a lower elevation than river level. This difference between tunnel level and river level is about 80 m as illustrated in Figure 4.

As shown in Figure 4, the overburden of each station is equal to 70 m, 130 m, and 240 m for LP, YB, and T2 stations, respectively. It may be expected that the principal stress orientations are affected by the magnitudes of different overburden stresses at the stations.



**Figure 4.** Comparison between tunnel and river level at the LP, YB, and T2 stations.

#### 4. Test equipment and methodology

The hydraulic fracturing test (HFT) and hydraulic test in pre-existing fracture (HTPF) at the Gotvand site were carried out using the main engineering equipment including:

- Three-piston electromotor pumps with controlled pressure capacity and flow of maximum 350 Pa and 1 m<sup>3</sup>/s, respectively.
- Hydraulic fracturing double packer with 95 mm diameter with the capacity of maximum pressure of 450 Pa.
- Ultrasonic flow meter
- Pressure sensor and data logger
- High pressure hoses and sealing rods capable of tolerating high water pressure
- Impression packer and pertinent items used to observe fractures and their orientation
- Borehole camera and compass
- Other accessories and tools such as container, winch, air compressor.

Some of the pieces of equipment used are shown in Figure 5.

The test procedure followed exactly the ASTM standard and ISRM suggested method (ASTM D4645, 1992; Haimson et al., 2003; Haimson and Cornet, 2003; Hanson, 2004).

Suitable test sections in each borehole were selected on the basis of the analysis of the rock core material and borehole logs. For HFT, homogeneous borehole sections without visible fractures were selected and, for HTPF tests, fractured sections mainly with different spatial orientation were selected.

After the installation of the equipment and the packer to the desired depth, the tests started with the following steps (Klee and Rummel, 2004):

- The packer inflated to a pressure of 49-59 Pa.



**Figure 5.** Some of the hydraulic fracturing test equipment. (a): Double packers; (b): Control room; (c) Borehole compass; (d): Main pump.

- The test interval pressurized rapidly to a pressure of about 20 Pa, and subsequently decayed for about 5 min in the case of HFT, and for about 15 min in the case of HTPF. Then interval pressure released. This is a permeability test, which is marked as first "P-test" in this study.
- The test interval pressurized with an injection rate ranging from 0.06 to 0.12 m<sup>3</sup>/s. The pressure inside of the test interval increases. This happens because the volume of the test interval before initiation of any fracture remains constant. In the case of HFT, increasing pressure continues until a fracture occurs inside the borehole, while in the case of HTPF this procedure continues up to a time of about 1 min.
- Injection and system shut-in terminated after about 3 min.
- The interval pressure released and the recovered fluid volume observed.
- In order to observe the shut-in pressure of fractures, several repetitions of re-pressurization of the test interval with constant injection rate of about 0.12-0.3 m<sup>3</sup>/s for about 1 min done.
- Injection and system shut-in terminated after about 3 min.
- The interval pressure released and the recovered fluid volume observed.

- A step-rate injection test with stepwise increase in the injection flow-rate conducted and the corresponding injection pressure observed.
- Injection and system shut-in terminated after about 3 min.
- The interval pressure released and the recovered fluid volume observed.
- The final step-rate test conducted as usually done for the classical Lugeon test, referred to in this study as "SP-test" (second permeability test).
- Finally, the packer deflated and prepared for the next test section.

## 5. Test results and discussion

### 5.1. Analysis and interpretation of HFT results

During each test, the injection pressure and injection flow rate and the recovered fluid volume were recorded by the digital data acquisition system. The derived parameters of HFT (breakdown pressure at the moment of fracture initiation,  $P_c$ ; fracture re-opening pressure,  $P_r$ ; shut-in pressure,  $P_{si}$ ; and the resulting in situ tensile strength,  $P_{co}$ ) as well as the results of the impression packer tests (fracture strike direction,  $\theta$ ; dip direction,  $\beta$ ; and dip,  $\alpha$ ) in the boreholes are summarized in Table 2. It is important to note that during the tests for the borehole no. LPHF2 the packer failed and the continuation of the test was interrupted; thus the complete data for LPHF2 are not available.

For instance, the analysis of the graphical test record at a depth of 36 m at the YB station, borehole no. YBHF1 is shown in Figure 6. In this diagram, the pressure-time plot and methods for determination of  $P_c$ ,  $P_r$ , and  $P_{co}$  are shown.

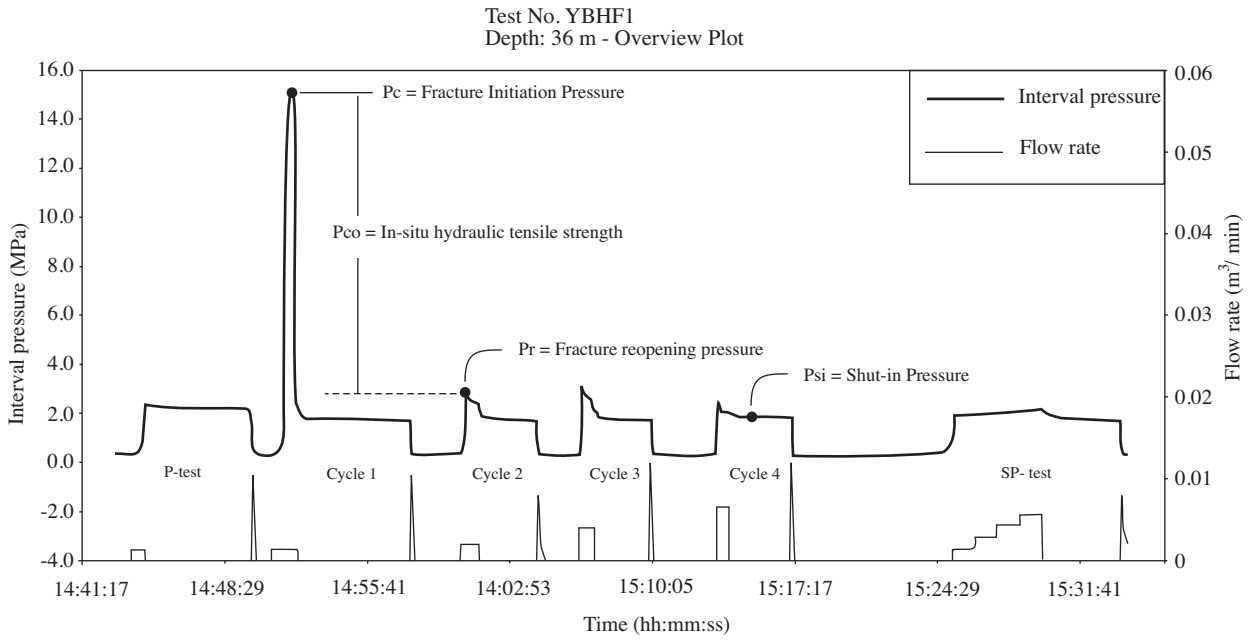
The shut-in pressure  $P_{si}$  was determined by the following 3 steps (Guo et al., 1993):

- The plot of pressure-injection flow-rate allows one to determine the exact pressure value at which the hydraulic flow terminates ( $Q = 0$ ). Therefore, the P-Q plot yields an upper-limit estimate of the shut-in pressure ( $P_{si(max)}$  in Table 2).
- A Muskat-type plot of the logarithm of the difference between the pressure  $P$  and an asymptotic pressure level  $P_a$  vs. time  $t$  yields the lower-limit of the shut-in pressure ( $P_{si(min)}$  in Table 2.), assuming that the linear part of the plot characterizes radial flow, i.e. the stimulated fracture is nearly closed.
- These 2 limits of shut-in pressure, which indicate acting stress normal to the fracture plane, mark a transition from a rapid linear pressure drop, observed immediately after shut-in, to the beginning of a diffusion-dominated slow pressure decrease. This transition can be determined by a tangent to the linear pressure decrease known inflection point in the pressure-time plot ( $P_{si(average)}$  in Table 2.). This approach was proposed by Gronseth and Kry (1981).

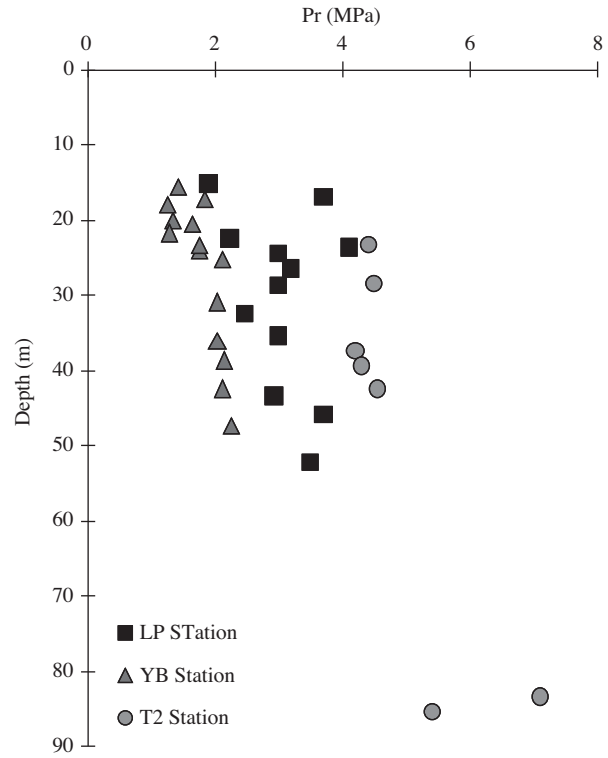
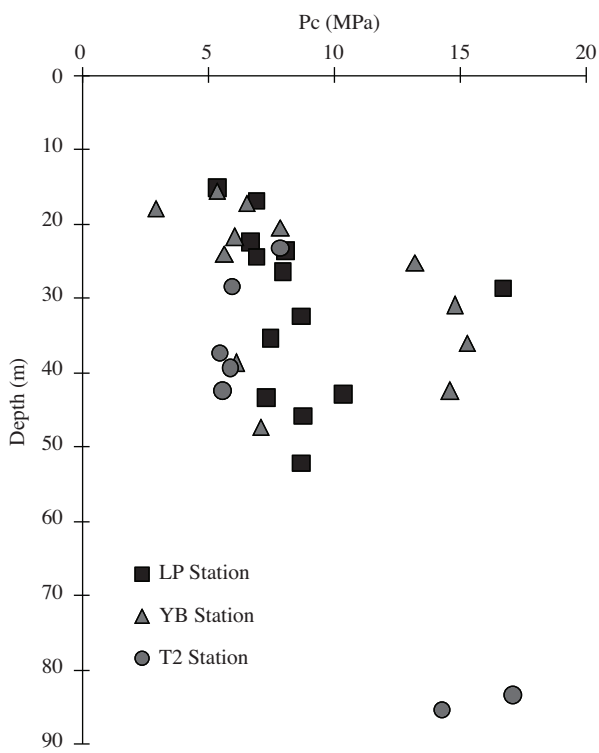
The derived characteristic pressure data ( $P_c$ ,  $P_r$ ,  $P_{si}$ ,  $P_{co}$ ) against true vertical depth (TVD), are shown graphically in Figure 7 to Figure 10, respectively. Note that the fracture initiation pressures of all the boreholes increased with increase in depth.

**Table 2.** Results of hydraulic injection tests in Gotvand Dam,  $P_c$ : fracture initiation pressure,  $P_r$ : fracture re-opening pressure,  $P_{co}$ : in situ hydraulic tensile strength,  $P_{si}$ : shut-in pressure,  $\theta$ : strike of fracture plane (north over east),  $\beta$ : dip direction of fracture plane (north over east),  $\alpha$ : dip of fracture plane (with respect to horizontal).

Station	Borehole	Material	Borehole length (m)	Pc (MPa)		Pr (MPa)	Pco (MPa)	Psi (MPa)			Orientation		
				HF	HFPPF			Average	Min	Max	$\theta$	$\beta$	$\alpha$
LP	LPHF1	Silt stone	26.5		7.97	3.18	4.74	2.50			30	120	75
		Silt stone	35.5		7.50	3.00	4.40	2.20					
		Silt stone	43.5		7.33	2.92	3.93	3.00					
		Sand stone	46.0		8.78	3.70	4.28	2.43					
		Silt stone	52.3		8.73	3.50	5.13	2.52					
	LPHF2	Silt stone	43.0		10.38	Failed	Failed	Failed	Failed	Failed			
	LPHF3	Clay stone	22.5		6.70	2.23	4.27	1.80			20	110	63
		Clay stone	24.5		6.94	3.00	3.69	2.27			30	120	20
	LPHF4	Clay stone	21.5		5.40	1.90	3.40	1.38			340	70	70
		Silt stone	24.0		6.95	3.70	3.02	2.36					
	LPHF5	Clay stone	46.1		8.73	2.46	6.03	2.43					
	LPHF6	Silt stone	33.5		8.10	4.10	3.98	2.73			10	100	0
Silt stone		40.7	16.75		3.00	13.25	2.04			40	130	55	
YB	YBHF1	Silt stone	31.0	14.77		2.03	12.47	1.93	1.79	1.96	30	120	83
		Silt stone	36.0	15.26		2.04	12.36	2.08	1.98	2.08	10	100	85
		Sand stone	38.8		6.12	2.14	3.72	2.35	2.14	2.36	48	318	20
		Clay stone	42.5	14.61		2.10	12.11	1.89	1.83	1.90	172	82	80
		Silt stone	47.3		7.08	2.25	4.63	2.23	2.08	2.23	5	275	15
	YBHF2	Sand stone	22.0		5.38	1.41	3.82	1.50	1.30	1.51	1	91	37
		Clay stone	24.5		6.53	1.83	4.43	1.53	1.24	1.61	28	118	82
		Silt stone	28.5			1.33		1.29	1.03	1.29	172	82	32
		Silt stone	30.9		6.09	1.27	4.49	1.35	1.19	1.34	5	95	33
		Silt stone	34.0		5.68	1.74	4.08	1.65	1.56	1.70	88	358	47
	YBHF3	Silt stone	25.4		2.92	1.25	1.43	1.17	1.09	1.22	15	285	66
		Silt stone	29.0		7.88	1.64	6.28	1.64	1.59	1.70	7	97	68
		Silt stone	32.8			1.75		1.75	1.65	1.75	88	358	65
		Silt stone	35.5	13.22		2.10	11.47	2.00	1.93	2.01	177	266	48
T2	T2HF1	Silt stone	23.4		7.9	4.4	3.9	2.8	2.3	2.4			
		Conglomerate	28.5		6.0	4.5	1.5	2.0	1.2	2.6			
		Conglomerate	37.5		5.5	4.2	1.3	2.3	1.0	2.8			
		Conglomerate	39.5		5.9	4.3	1.6	2.4	1.1	2.5			
		Conglomerate	42.5		5.6	4.6	1.1	3.0	1.2	3.0			
		Sand stone	83.5	17.1		7.1	10	4.3	2.9	4.8	30	120	90
		Sand stone	85.5	14.3		5.4	8.9	3.8	3.0	3.9	60	150	85
		Sand stone	91.6	11.3		4.5	6.8	3.8	3.8	4.1			



**Figure 6.** Graphical diagram of HFT result at the YB station at Gotvand Dam site (borehole no. YBHF1-depth: 36 m).



**Figure 7.** Breakdown pressure Pc at Gotvand Dam site.

**Figure 8.** Fracture re-opening pressure (Pr) at Gotvand Dam site.



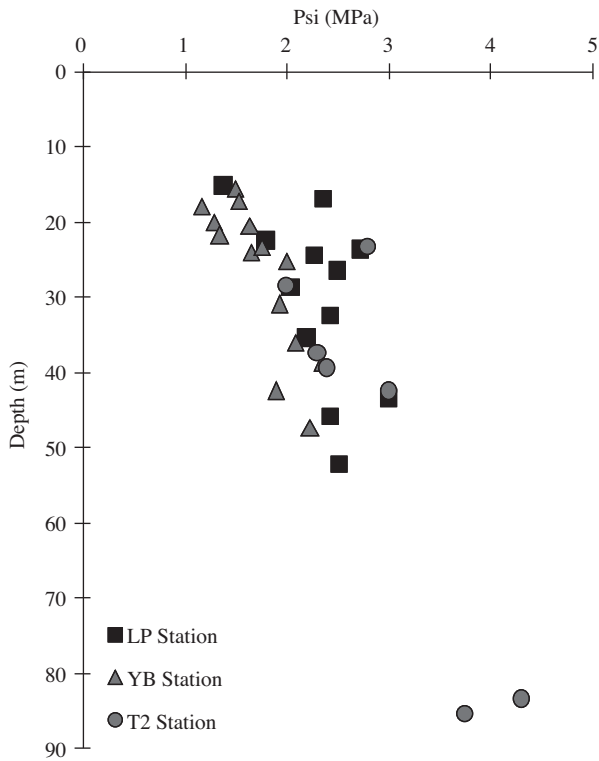


Figure 9. Shut-in pressure (Psi) at Gotvand Dam site.

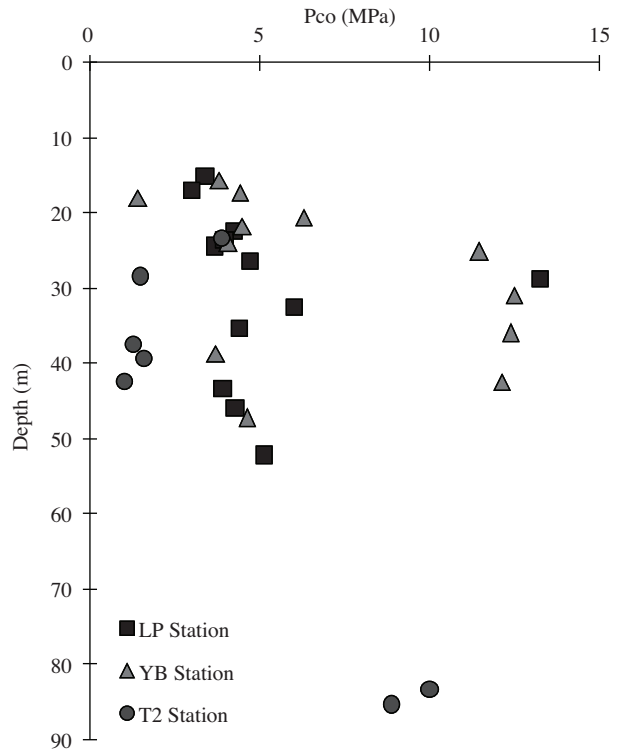


Figure 10. In situ tensile strength (Pco) at Gotvand Dam site.

### 5.2. Tracing of the fracture

After completion of all HFT and HTPF, the straddle packer tool was replaced by the impression packer. The impression packer tests consisted of an inflation of the impression packer element to a pressure approximately 20% above the fracture re-opening pressure for a time of about 30 min. After recovery of the packer tool to the surface, the fracture trace is marked on the packer sleeve and reflected to a transparent plastic cover sheet wrapped around the packer. The captured pictures from the borehole camera and compass unit provide the orientation of the reference mark with respect to magnetic north. Examples of the fracture traces observed during impression packer testing are shown in Figure 11.



Figure 11. Effect of HF crack on impression packer plastic cover at YB station of Gotvand.

Using the above method, all the hydraulic fracturing tests and hydraulic injection tests in pre-existing fractures as well as impression packer tests for fracture orientation determination were carried out in the AJ and BK formation boreholes.

### 5.3. Stress evaluation from hydraulic fracturing tests

To derive stresses from measurements in boreholes, it is necessary to consider the stress distribution around a circular hole subjected to far-field compressive stresses. This solution was first derived by Kirsch (1898). In the case where the borehole axis is parallel to the maximum principal stress, the Hubbert and Willis formula for the critical pressure  $P_c$  at the moment of fracture initiation is used (Hubbert and Willis, 1957; Haimson and Fairhurst, 1970):

$$P_c = 3.S_h - S_H + P_{co} - P_p \quad (1)$$

where  $S_h$  and  $S_H$  are the horizontal far-field principal stresses,  $P_{co}$  is the in situ tensile strength of the rock, and  $P_p$  is the pore pressure in the rock mass.  $P_c$  is often called the breakdown pressure during the hydraulic fracturing process. It is assumed that the vertical stress is a principal stress and equal to the overburden stress, and the rock is homogeneous, isotropic, and initially impermeable, and that the induced fracture is oriented perpendicular to the minimum horizontal principal stress  $S_h$ . The last assumption yields:

$$P_{si} = S_h \quad (2)$$

where  $P_{si}$  is the shut-in pressure to merely keep the fracture open after the pressurizing system. After pressure release, the fracture may close. It can be re-opened during subsequent pressure cycles at a pressure of:

$$P_r = 3.S_h - S_H - P_p = P_c - P_{co} \quad (3)$$

Using this linear elastic approach, the principal stresses can be expressed by the relations:

$$\begin{aligned} S_h &= P_{si} \\ S_H &= 3.P_{si} - P_r - P_p \\ S_v &= \rho.g.z \\ P_{co} &= P_c - P_r \end{aligned} \quad (4)$$

Thus, the stress analysis only requires knowledge of the rock mass density,  $\rho$ , the determination of characteristic pressure values in particular the shut-in pressure,  $P_{si}$ , and re-opening pressure,  $P_r$ , at depth  $z$  where the fracture is produced.

Considering the simple and idealistic assumptions used in the above approach, the determination of stresses by Eqs. (4) may sometimes be questioned. This, in particular, applies to the assumptions on rock mass isotropy.

The resulting minimum and intermediate principal stresses,  $S_h$  and  $S_H$ , are listed in Tables 3 to 5, and are shown graphically in Figures 12 to 14, for stations LP, YB, and T2, respectively. For vertical stress calculation an average value of about  $2.480 \text{ g/cm}^3$  was considered, according to the results of experiments on the rock density.

**Table 3.** Results of the stress evaluation using the Hubbert and Willis (1957) approach at the LP station, Gotvand Dam.

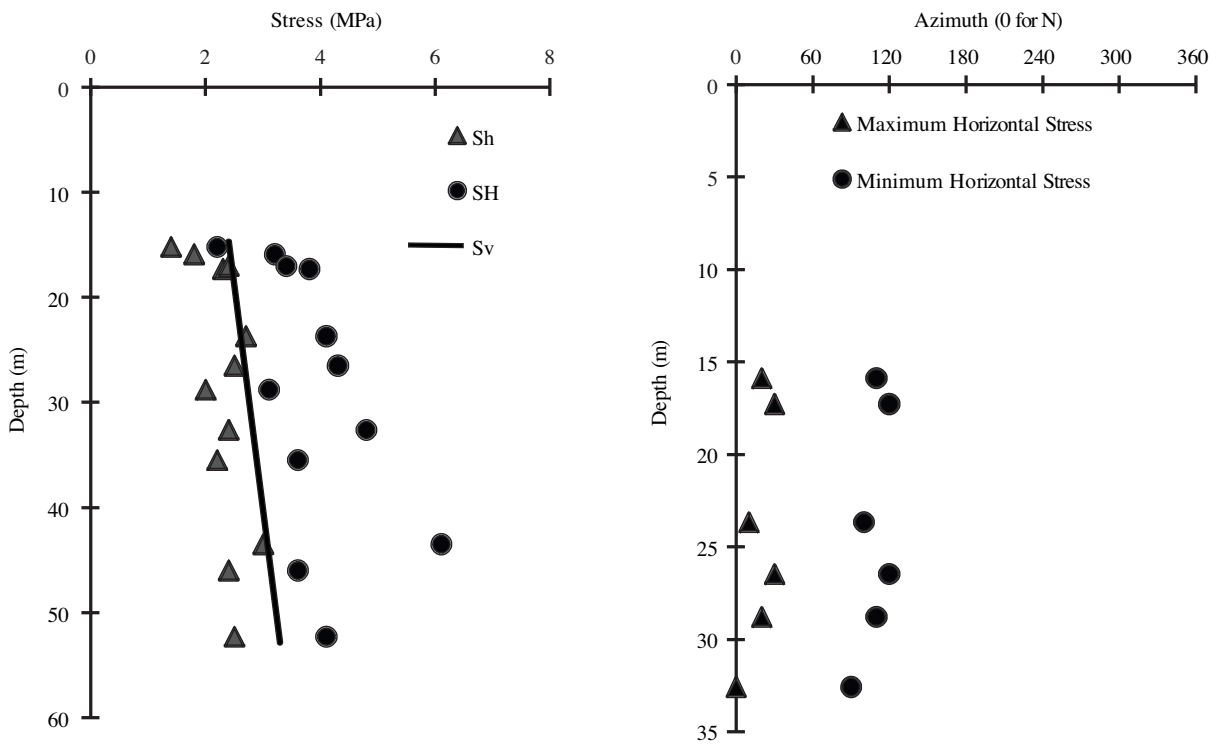
Borehole no.	Test code	Borehole length (m)	Sh (MPa)	SH (MPa)	SV (MPa)	$\theta$ for SH (deg)
LPHF1	LPHF1-26.5	26.5	2.5	4.3	2.6	N30E
	LPHF1-35.5	35.5	2.2	3.6	2.9	-
	LPHF1-43.5	43.5	3.0	6.1	3.1	-
	LPHF1-46	46.0	2.4	3.6	3.1	-
	LPHF1-52.25	52.3	2.5	4.1	3.3	-
LPHF3	LPHF3-22.5	22.5	1.8	3.2	2.5	N20E
	LPHF3-24.7	24.7	2.3	3.8	2.6	N30E
LPHF4	LPHF4-21.5	21.5	1.4	2.2	2.4	-
	LPHF4-24	24.0	2.4	3.4	2.4	-
LPHF5	LPHF5-46.1	46.1	2.4	4.8	2.8	N0E
LPHF6	LPHF6-33.5	33.5	2.7	4.1	2.6	N10E
	LPHF6-40.7	40.7	2.0	3.1	2.7	N20E

**Table 4.** Results of the stress evaluation using the Hubbert and Willis (1957) approach at the YB station, Gotvand Dam.

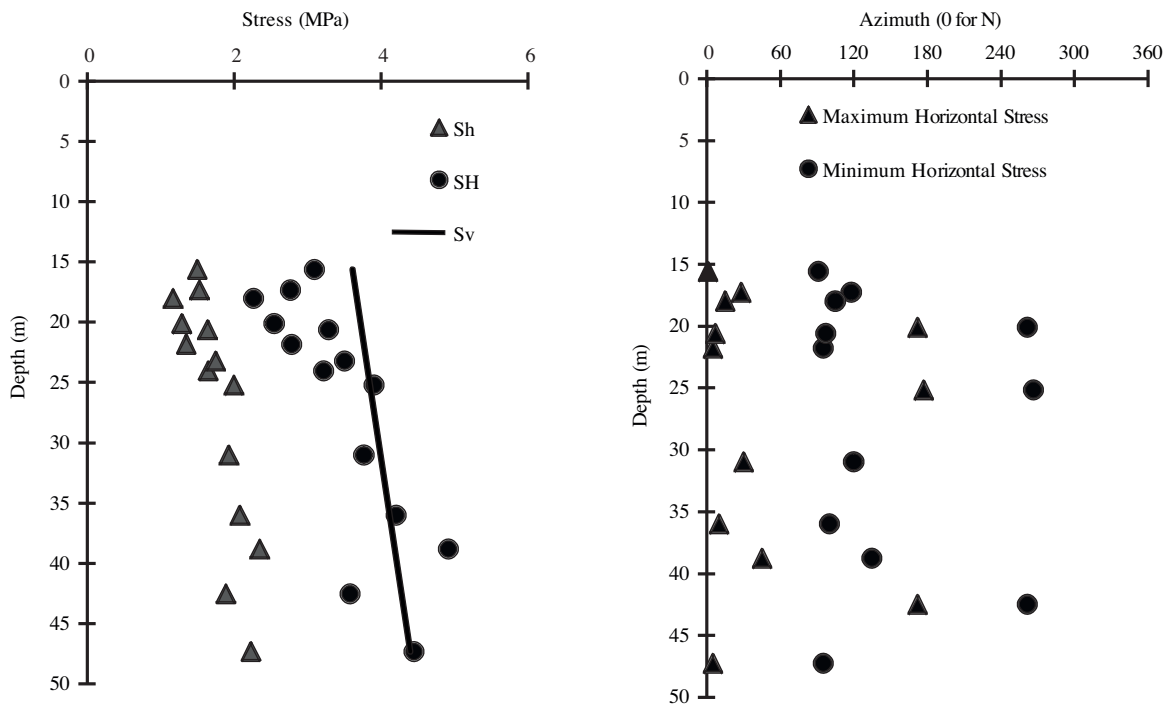
Borehole no.	Test code	Borehole length (m)	Sh (MPa)	SH (MPa)	SV (MPa)	$\theta$ for SH (deg)
YBHF1	YBHF1-d31	31	1.93	3.76	3.9928	N30E
	YBHF1-d36	36	2.08	4.2	4.1168	N103
	YBHF1-d38.8	38.8	2.35	4.91	4.1862	N48E
	YBHF1-d42.5	42.5	1.89	3.57	4.278	N172E
	YBHF1-d47.3	47.3	2.23	4.44	4.397	N5E
YBHF2	YBHF2-d22	22	1.5	3.09	3.6109	N1E
	YBHF2-d24.5	24.5	1.53	2.76	3.653	N28E
	YBHF2-d28.5	28.5	1.29	2.54	3.7225	N172E
	YBHF2-d30.9	30.9	1.35	2.78	3.7646	N5E
	YBHF2-d34	34	1.65	3.21	3.8192	-
YBHF3	YBHF3-d25.4	25.4	1.17	2.26	3.6704	N15E
	YBHF3-d29	29	1.64	3.28	3.7349	N7E
	YBHF3-d32.8	32.8	1.75	3.5	3.7994	-
	YBHF3-d35.5	35.5	2	3.9	3.849	N177E

**Table 5.** Results of the stress evaluation using the Hubbert and Willis (1957) approach at the T2 station, Gotvand Dam.

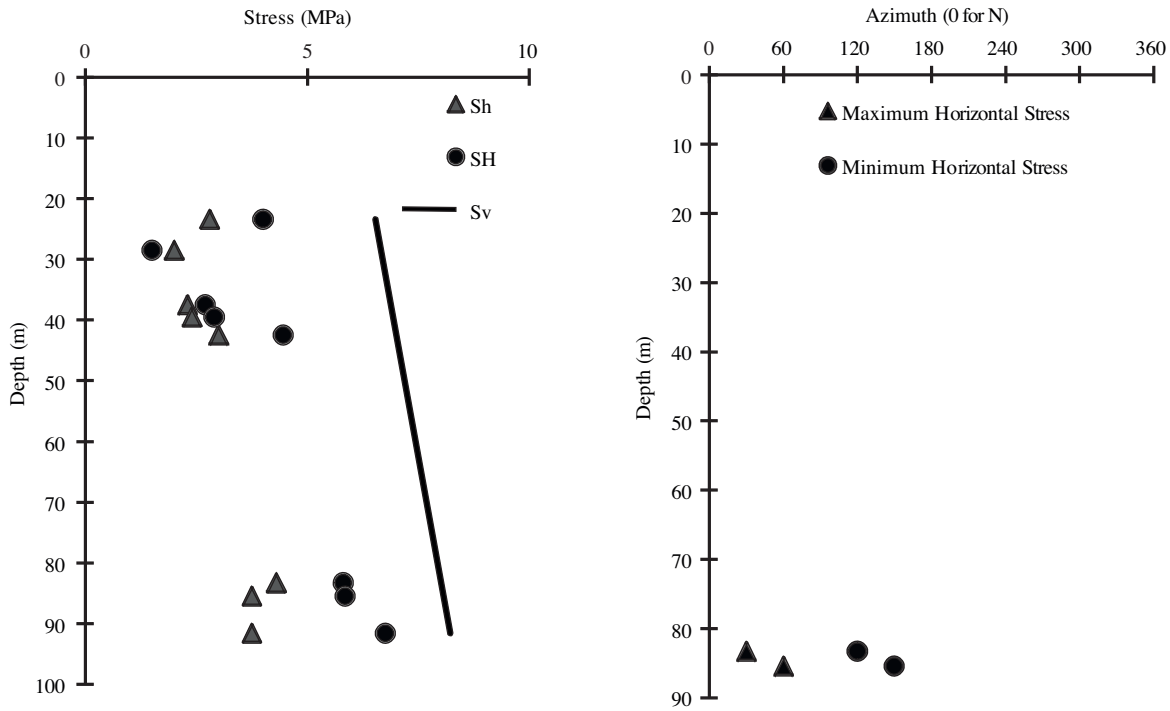
Borehole no.	Test code	Borehole length (m)	Sh (MPa)	SH (MPa)	SV (MPa)	$\theta$ for SH (deg)
T2-1	T2-1-23.4	23.4	2.8	4	6.532	-
	T2-1-28.5	28.5	2	1.5	6.659	-
	T2-1-37.5	37.5	2.3	2.7	6.882	-
	T2-1-39.5	39.5	2.4	2.9	6.932	-
	T2-1-42.5	42.5	3	4.45	7.006	-
	T2-1-83.3	83.3	4.3	5.8	8.018	N30E
	T2-1-85.6	85.5	3.75	5.85	8.075	N60E
T2-1-91.6	91.6	3.75	6.75	8.224	-	



**Figure 12.** Principal stress interpretations at the LP station: (a) Principal stress vs. depth; (b) Horizontal stress orientation vs. depth



**Figure 13.** Principal stress interpretations at the YB station: (a) Principal stress vs. depth; (b) Principal stress orientation vs. depth.



**Figure 14.** Principal stress interpretations at the T2 station: (a) Principal stress vs. depth; (b) Principal stress orientation vs. depth.

Because of the lower elevation of the LP station with respect to the river level and less overburden compared to the other stations, a horizontal stress is a maximum principal stress caused by tectonic effects. In contrast, at the YB and T2 stations, the vertical stress component is the maximum principal stress. In the other words, this study shows that at the lower penstock tunnels (LP station) the vertical stress,  $\sigma_v$ , due to the small weight of the overburden with given rock density is the intermediate principal stress, while in the tunnels located at the YB and T2 stations,  $\sigma_v$  is the maximum principal stress.

#### 5.4. Maximum horizontal principal stresses orientation

The hydraulic fracturing tests were conducted in both fractured and intact rocks. Regarding the fracture planes being induced, the impression packer is used for determining the actual principal stresses. As shown in Table 2., the measurements yield an orientation of the maximum horizontal stress, which is the same as most of the induced and pre-existing fractures, with strike direction of  $N 30^\circ \pm 5^\circ$  (NE-SW). This direction is generally the same as directions indicated by focal mechanisms solutions from earthquakes (Figure 15). As shown in Figure 16, the strike direction of the maximum horizontal principal stress is obtained perpendicular to the general fault extension in the test location.

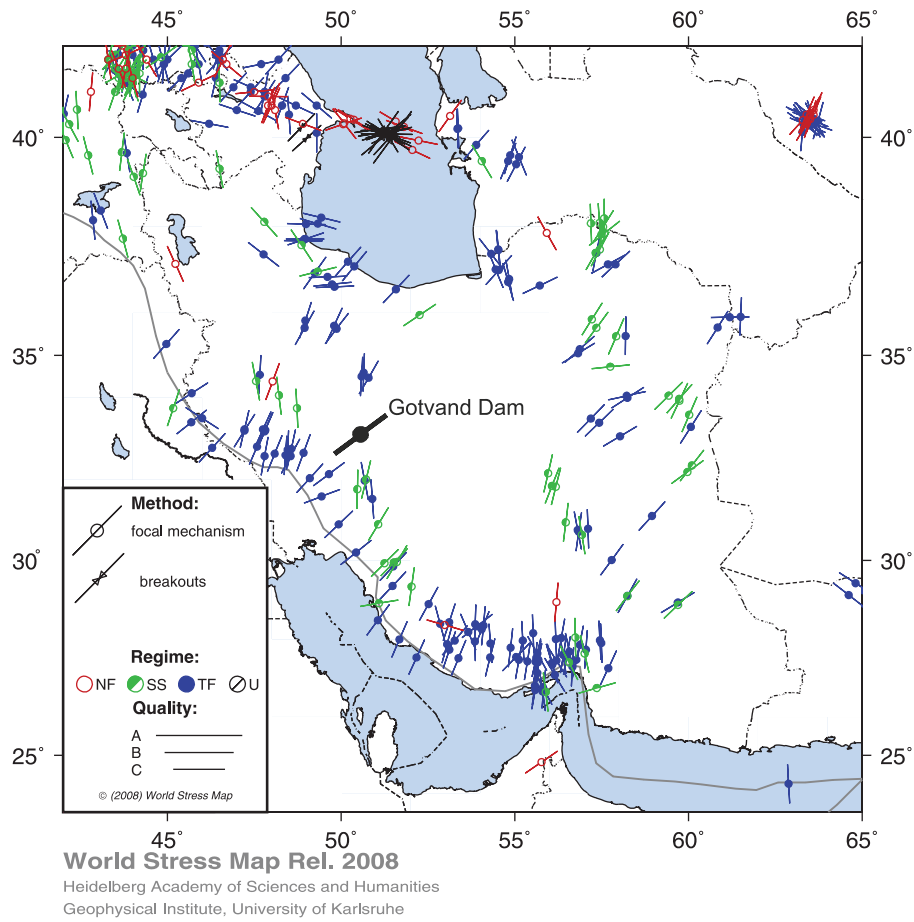


Figure 15. Directions indicated by HFT on Iranian focal mechanism solutions from earthquake map.

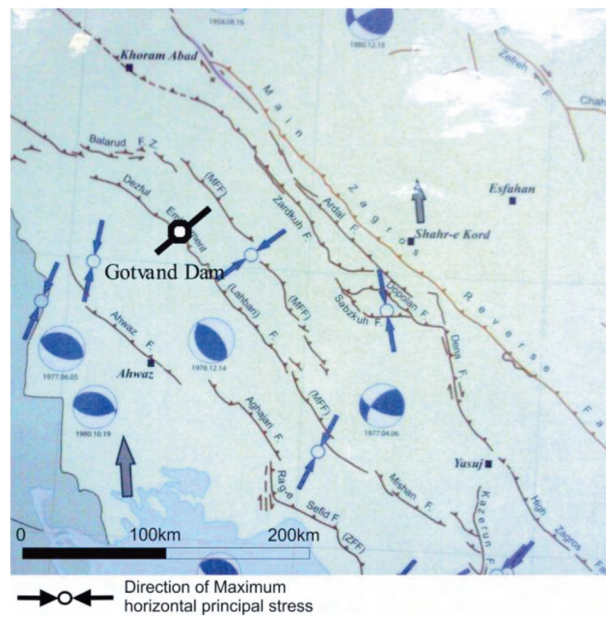


Figure 16. Directions indicated by HFT on Iranian fault system.

## 6. Conclusions

The HFT results at the Gotvand Dam site can be summarized as follows:

- Results of all hydraulic fracturing tests showed that all of the HFT and HTPF parameters ( $P_c$ ,  $P_r$ ,  $P_{si}$ , and  $P_{co}$ ) increase with depth.
- To evaluate the maximum, intermediate, and minimum principal stresses, the Hubbert and Willis formula was used. The principal stress interpretations showed that due to the small weight of the overburden with given rock density at the LP station the vertical component of stress is the intermediate principal stress and a horizontal stress is a maximum principal stress caused by tectonic effects. In contrast, at the other stations, because of such high overburden, the vertical stress component is the maximum principal stress.
- The measurements yield an orientation of the maximum horizontal stress with strike direction of  $N 30^\circ \pm 5^\circ$  (NE-SW). This direction is generally compatible with the directions indicated by focal mechanisms solutions from earthquakes. It is also found that the strike direction of most of the induced and pre-existing fractures are the same as the general fault extension in the test location.

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