

1 **1. Introduction**

2 In recent years, human induced seismicity associated with underground wastewater
3 disposal and fluid injection has become a matter of societal concern. Seismicity rates
4 have increased dramatically in regions far from active tectonic margins, and stable
5 continental regions like the Western Canada Sedimentary basin (e.g. Atkinson et al.,
6 2016; Bao and Eaton, 2016) and the central United States (e.g. Keranen et al., 2013;
7 Frohlich and Brunt, 2013; Ellsworth, 2013; Langenbruch and Zoback, 2016) have seen
8 sharp increases of moderate to large earthquakes, with $M_w > 5$ events becoming
9 common. In Europe, induced earthquakes during fluid pressure stimulation of subsurface
10 reservoirs have been documented in several notable cases including Switzerland
11 (Deichmann and Giardini, 2009), southern Italy (Improta et al., 2009) and the
12 Netherlands (van Thienen-Visser and Breunese, 2015).

13 Within plate interiors, surveys of crustal stress and measurements from deep
14 boreholes have shown that the crust is critically stressed, with shear stress levels near the
15 strength limit for brittle failure (Townend and Zoback 2000). Under these conditions, the
16 maximum stress level that can be supported is limited by the frictional strength of pre-
17 existing ancient faults. Thus, even small changes in the stress field surrounding ancient
18 faults can trigger earthquakes (Stein, 1999) (Fig. 1a). It has long been known that
19 underground fluid injection can induce seismicity (e.g., Raleigh et al., 1976; Simpson et
20 al., 1988). Long-term fluid injection at high rates nearby pre-existing faults can modify
21 the surrounding stress field (either directly or indirectly) causing reactivation of
22 preexisting faults (e.g., Ellsworth, 2013). The basic physical mechanism for inducing
23 seismicity is well understood in terms of the effective stress principle (Hubert and Rubey,

24 1959; Sibson, 1986):

$$25 \quad \tau = C + \mu(\sigma_n - P_f) \quad (1)$$

26 where τ is the shear stress acting on the fault, C is cohesion, and μ is the coefficient of
27 friction which is multiplied by the difference between the normal stress (σ_n) and fluid
28 pressure (P_f), which represents the effective normal stress (σ'_n). During underground
29 fluid injection, propagation of a fluid pressure front from the injection point reduces the
30 effective normal stress acting on incipient fault planes, promoting earthquake failure,
31 with larger events expected for higher rates and longer periods of pumping (e.g. Hubert
32 and Rubey, 1959; Shapiro et al., 2003; Keranen et al., 2014; McGarr, 2014; Bao and
33 Eaton, 2016) (Fig. 1a and b).

34 The Coulomb failure relation of Equation 1 predicts the stress conditions for fault
35 slip (Fig. 1b) but it does not address the question of frictional stability and whether slip
36 will be seismic or aseismic upon reactivation. The stability of frictional sliding is
37 determined by the local elastic stiffness around the fault and the fault zone friction
38 constitutive properties (Rice and Ruina, 1983). Rate- and state- frictional (RSF)
39 constitutive equations are commonly employed to describe fault friction and the resulting
40 slip behavior (Dieterich, 1979; Ruina, 1983; Marone, 1998):

$$41 \quad \mu = \mu_0 + a \ln\left(\frac{v}{v_0}\right) + b \ln\left(\frac{\theta v_0}{D_c}\right) \quad (2)$$

42 where, upon a velocity increase from v_0 to v , the coefficient of friction (μ) suddenly
43 increases (direct effect, a) from a reference steady state (μ_0) and then evolves to a new
44 steady state (evolution effect, b) over a characteristic critical slip distance (D_c) (Fig. 1c).

45 The state variable, θ is commonly interpreted as the average lifetime of frictional contacts
46 and it evolves over the critical slip distance D_c following a state evolution law such as

47 (Ruina, 1983; Marone, 1998):

48
$$\frac{d\theta}{dt} = -\frac{v\theta}{D_c} \ln\left(\frac{v\theta}{D_c}\right) \quad (3)$$

49 Under conditions of steady state shear $d\theta/dt=0$ and $\theta_{ss}= D_c/v$. The dependence of
50 frictional strength on slip rate is described by the friction rate parameter $(a-b) =$
51 $\Delta\mu_{ss}/\log(v/v_0)$. If friction increases with increasing velocity, $(a-b)>0$, the material is said
52 to be velocity strengthening and slip is inherently stable, leading to aseismic fault creep
53 (Fig. 1c). However, if the material is velocity weakening, $(a-b)<0$, frictional strength
54 decreases with slip velocity and slip may be unstable, satisfying the conditions for the
55 nucleation of a seismic instability, depending on the rate of weakening with slip $(b-a)/D_c$.

56 Combining elastic dislocation theory with RSF constitutive equations provides a
57 general description for the criterion of fault stability (Ruina, 1983; Gu et al., 1984). For a
58 velocity weakening fault gouge, a dynamic frictional instability will nucleate when the
59 stiffness of the loading system, k , is lower than a critical fault rheologic stiffness, k_c ,
60 defined by:

61
$$k_c = (\sigma_n - P_f)(b - a) / D_c \quad (4)$$

62 Equation 4 shows that an increase in fluid pressure reduces k_c , favoring stable
63 sliding rather than earthquake slip (Fig. 1a). This prediction contrasts with seismological
64 observations that show a strong link between massive fluid injection and induced
65 seismicity. In addition, Equation 4 predicts earthquake slip only if the fault has a velocity
66 weakening behavior, i.e. $(b-a)$ positive, while laboratory experiments show that at
67 stress/temperature conditions typical of the occurrence of induced seismicity, i.e. < 5 km,
68 a wide variety of fault gouge materials show velocity strengthening frictional behavior
69 (e.g. Blanpied et al., 1998; Ikari et al., 2011; Samuelson and Spiers, 2012; Scuderi et al.,

70 2013; Kohli and Zoback, 2013; Scuderi and Collettini, 2016;).

71 This conundrum suggests some gaps in our understanding of induced seismicity
72 and the physical processes governing fault slip under overpressurized fluid conditions.
73 The purpose of this paper is to improve on our understanding of induced seismicity and,
74 thereby, to improve our ability to evaluate the seismic risk associated with human
75 induced earthquakes.

76 Resolving these apparent inconsistencies and developing valid predictive models
77 for earthquakes induced by fluid injection remain important challenges. To address this
78 issue, we developed laboratory experiments reproducing the boundary conditions of
79 induced seismicity along ancient faults, where the tectonic shear stress is nearly constant
80 (Townend and Zoback, 2000) and fluid pressurization results in a systematic reduction of
81 the effective normal stress.

82 **2. Materials and Methods**

83 We performed laboratory experiments using a biaxial apparatus, BRAVA
84 (Collettini et al., 2014), in a double-direct shear configuration (DDS) within a pressure
85 vessel to allow a true-triaxial stress field (Fig. 2a). In this configuration, two fast acting
86 servo-controlled rams are used to apply normal (σ_n) and shear stress (τ) to the fault zones.
87 Each ram can be controlled either in load-feedback mode, to maintain a constant load, or
88 in displacement-feedback mode, in which case the ram is advanced at a constant
89 displacement rate. Forces were measured using strain-gauged hollow load cells
90 (manufactured by LEANE International model CCDG-0.1-100-SPEC), positioned inside
91 the pressure vessel, with an amplified output of ± 5 V for a maximum force of 1.5 MN
92 and an accuracy of ± 0.01 kN, which are calibrated regularly. Displacements were

93 measured via Linear Variable Differential Transformers (LVDTs), referenced at the load
94 frame and the moving ram, with an accuracy of $\pm 0.01 \mu\text{m}$. Load point displacement
95 measurements are corrected for the stiffness of the testing apparatus, with nominal values
96 of 386.12 kN/mm for the vertical frame and 329.5 kN/mm for the horizontal frame. In
97 this configuration, the horizontal LVDT measures the evolution of gouge layer thickness
98 that we corrected for the geometrical layer thinning associated with the DDS geometry
99 (Scott et al., 1994). Confining pressure (P_c) and up- and down-stream pore fluid pressure
100 (P_{pu} and P_{pd} respectively) were applied using three hydraulic fast-acting servo-controlled
101 intensifiers (Fig. 2a). Displacements were measured via LVDTs and pressures were
102 monitored with diaphragm pressure transducers accurate to ± 7 kPa. Confining pressure
103 was applied using a hydrogenated paraffinic white oil (XCEL THERM 600, Radco
104 Industries), and maintained constant throughout each test using a load-feedback control
105 loop. Pore fluid pressure was applied using a calcium rich water solution similar to the
106 water circulating in carbonate bearing faults. Output signals were digitalized using a
107 simultaneous multichannel analog to digital converter with 24-bit/channel resolution at a
108 sampling rate of 10 kHz, and then averaged for storage at rates between 1 Hz and 10 kHz.

109 Our double-direct shear configuration consists of three stainless steel forcing
110 blocks that confine and shear two layers of simulated fault gouge (Fig. 2b). The steel
111 blocks are equipped with conduits to allow fluid flow and connect the gouge layers with
112 the pore fluid intensifiers. Sintered porous frits (permeability $\sim 10^{-14} \text{ m}^2$) are press fit in
113 cavities within the forcing blocks to allow a homogenous distribution of fluids on the
114 entire sample surface, and are equipped with grooves, 0.8 mm in height with 1 mm
115 spacing, to ensure shear localization within the fault gouge and not at the layer

116 boundaries. The nominal frictional contact area is 5.54 cm × 5.55 cm, and we refer all
117 measurements of stress, displacement and pressure changes to one layer. For these
118 sample dimensions and loading configuration, normal stress on the gouge layers is
119 determined by the summation of applied stress (σ_n) and confining pressure (P_c), with the
120 effective normal stress acting on the gouge layers given by: $\sigma'_n = (\sigma_n + P_c) - P_f$.

121 We simulate fault gouge using granular powders of Carrara marble with a grain
122 size <125 μm and a composition of >98% CaCO_3 . In laboratory experiments, granular
123 powders are used as analogues for fault gouge material and Carrara marble is commonly
124 used as an analog of carbonate bearing fault zones (e.g., Verberne et al., 2014; Carpenter
125 et al., 2016). Gouge layers were constructed using leveling jigs to obtain a uniform layer
126 thickness of 5 mm for all experiments. To ensure that each experiment started at similar
127 porosity, we weighted the gouge layers during and after (i.e. steel blocks + gouge
128 material) sample construction. We ensured that both layers had the same weight for each
129 experiment (Table 1). Using this procedure, we obtained variability < 6% in initial
130 sample density. Subsequently, the sample assembly was jacketed to separate the gouge
131 layers and pore fluids from the confining oil (Fig. 2b and details in Scuderi and Collettini,
132 2016).

133 **2.1 Experiment design and loading boundary conditions.**

134 We performed two types of experiments: 1) constant displacement rate
135 experiments to determine fault zone strength and permeability, and 2) creep experiments
136 to evaluate the evolution of slip behavior as a function of fluid overpressure. Both types
137 of experiments followed a common loading up procedure for comparison and
138 reproducibility purposes. We started by applying the confining pressure in steps of 1 MPa

139 every 5 minutes to allow for sample compaction until the target was reached. The applied
140 normal stress was then increased to the target value and maintained constant throughout
141 the experiment. At this stage, the up-stream pore fluid intensifier was advanced to apply
142 a small pore fluid pressure, generally 1MPa, while the down-stream intensifier was left
143 open to the atmosphere until flow through the gouge layer was established. Once we
144 ensured that gouge layers were fully saturated and all the residual air in the gouge was
145 expelled, the down-stream intensifier was closed to the atmosphere, and left to equilibrate
146 with the P_{pu} . Pore fluid pressure was then increased in steps of 1 MPa every 5 minutes to
147 the target value. The sample was left to equilibrate for about 30 minutes while creep
148 compaction occurred and the layer reached a steady state thickness. Shearing began at
149 this point, once the gouge particles had reached a close packing configuration. All
150 experiments were performed under nominally drained boundary conditions of constant
151 P_f .

152 **2.2 Fault strength and permeability.**

153 We conducted experiments at constant σ'_n of 10, 15 and 20 MPa and under
154 hydrostatic boundary conditions (i.e. $\lambda = P_f / \sigma_n = 0.4$) (Table 1). Shear stress was applied at
155 constant displacement rate of 10 $\mu\text{m/s}$ until the steady state strength was achieved. At this
156 point we stopped the vertical ram and measured fault zone permeability under quasi-static
157 loading conditions (note that creep occurs for the constant stress boundary conditions).
158 Permeability was measured using a constant head method that consists of imposing a
159 differential pressure (usually 1MPa) between the up- and down-stream fluid intensifiers
160 and measure the resulting flow rate across the gouge layers. We calculated permeability
161 using Darcy's law:

162
$$k = \frac{Q}{A\eta} \frac{dl}{dp_p} \quad (5)$$

163 where k is the sample permeability [m^2], Q is the measured flow rate [m^3s^{-1}], A is the
164 cross-sectional area [m^2], η is the viscosity of water [$MPa \cdot s$], ΔP_p is the imposed
165 differential pore pressure [MPa], and dl is the sample thickness. We assume $\eta =$
166 $1.002 \times 10^{-9} MPa \cdot s^{-1}$, define dl from the initial, measured layer thickness and changes
167 recorded by the LVDT on the horizontal piston, and Q as the average value of the flow
168 rates measured at the up-stream (P_{pu}) and down-stream (P_{pd}) pumps. To ensure steady
169 state flow conditions, we always waited until the flow rate difference, between Q_u and
170 Q_d , was less than 5%.

171 At the end of the permeability test the vertical piston was retracted until the shear
172 load was null. We increased the normal stress and the pore fluid pressure to achieve the
173 next σ'_n target, and repeated the procedure explained above.

174 **2.3 Creep experiments.**

175 Each creep experiment began at effective normal stress of 20 MPa and under hydrostatic
176 pore fluid pressure conditions (i.e. $\lambda=0.4$). Shear stress was applied by advancing the
177 vertical ram at constant displacement rate of 10 $\mu m/s$ for ~ 13 mm to achieve a steady
178 state shear strength (τ_{ss}) and ensure shear localization within the gouge layers (Fig. 3a
179 and Table 1). Next, we stopped the vertical ram and let the sample relax for 30 minutes,
180 to ensure crack closure and closest packing configuration within the sample. At this
181 stage, we started the creep test by switching the control of the vertical ram from
182 displacement-mode to load-mode, to maintain a constant shear stress on the gouge layers.
183 In creep mode, we measure the resulting fault slip at a given shear load and effective

184 normal stress. We set the shear stress at either 80% or 90% of the steady state shear
185 strength τ_{ss} (Fig. 3a and Table 1). Samples were left to deform under these boundary
186 conditions for 1 hr before fluid injection began. Fluids were injected by increasing the
187 pore fluid pressure stepwise from the up-stream intensifier, with fluid circulation and
188 equilibration modulated by the permeability of the fault, and following two similar but
189 different protocols: 1) we increased P_f by 1 MPa every hour or, 2) we increased P_f by 0.2
190 MPa every 12 minutes (Fig. 3a). We also performed experiments where the sample was
191 left to creep under hydrostatic boundary conditions for ~ 12 hr to monitor fault creep in
192 the absence of fluid pressurization.

193 **3. Results**

194 **3.1 Short term strength and fault permeability.**

195 We measured the frictional shear strength for steady-state sliding, τ_{ss} , under a
196 range of conditions (Table 1). As expected, τ_{ss} scales linearly with effective normal stress
197 according to the Coulomb-Mohr failure relation (Equation 1). The linear relationship
198 between effective normal stress and shear stress yielded a cohesion of 0.19 MPa and a
199 value of $\mu_{ss}=0.55$ (Fig. 3b), in agreement with previous works on Carrara marble (e.g.
200 Verberne et al., 2015; Carpenter et al., 2016). These results were reproducible across
201 multiple experiments (Table 1) with values of τ_{ss} varying between 5.7 and 5.5 MPa at σ'_n
202 = 10 MPa, 8.2 and 8.4 MPa at $\sigma'_n = 15$ MPa, and 11.6 and 11.8 at $\sigma'_n = 20$ MPa.

203 For each effective normal stress, we sheared layers until they reached a stable
204 friction value, which is associated with a steady-state shear fabric (e.g., Marone, 1998)
205 and then measured fault zone permeability. Permeability decreased with increasing
206 effective normal stress, with values of $5 \times 10^{-17} \text{m}^2$ at $\sigma'_n = 10$ MPa, $1.5 \times 10^{-17} \text{m}^2$ at $\sigma'_n =$

207 15 MPa and $7 \times 10^{-18} \text{m}^2$ at $\sigma'_n = 20 \text{ MPa}$ (Fig. 3b). Permeability values in the range of
208 $\sim 10^{-17} \text{m}^2$ facilitate fluid movement, suggesting that the experimental fault zone is under
209 fully drained boundary conditions (e.g. Townend and Zoback, 2000).

210 **3.2 Creep Behaviour.**

211 To evaluate fault stability during fluid pressurization we maintained a constant
212 shear stress and increased pore fluid pressure (Fig. 3b) while monitoring fault slip. The
213 evolution of fault slip shows the typical trimodal creep behavior described for creep of
214 intact rocks (e.g., Brantut et al., 2013), characterized by: (1) a primary or decelerating
215 creep, (2) a secondary or steady state creep and, (3) a tertiary creep where fault zone
216 acceleration culminates with dynamic failure (Fig. 4 and 5).

217 For the experiments performed at 90% of τ_{ss} the primary creep stage was limited
218 to the first 40 minutes of the test, during which the fault accumulated a displacement of
219 $\sim 30 \mu\text{m}$ (Fig. 4). In the experiments at 80% of τ_{ss} the period of primary creep was shorter,
220 with duration of ~ 30 minutes, during which the fault slipped up to $\sim 10 \mu\text{m}$ (Fig. 5). We
221 note that for our loading procedure, fluid injection always began at the end of primary
222 creep.

223 The secondary creep phase is characterized by a quasi-linear evolution of slip
224 with time as shown by the linear fit performed to retrieve creep velocity (Fig. 4 and 5).
225 For the case of 90% of τ_{ss} , the experiments performed at constant pore fluid pressure
226 show creep velocity of 16 nm/s, which corresponds to a shear strain rate ($\dot{\gamma}$) of $7 \times 10^{-5} \text{s}^{-1}$
227 (Fig. 4a and b). When the shear stress was 80% of τ_{ss} we document creep velocity of 5
228 nm/s corresponding to shear strain rate of $2.3 \times 10^{-5} \text{s}^{-1}$ (Fig. 5a and b). These values

229 represent the creep rates under constant fluid pressure conditions. For the ~12 hr duration
230 of these experiments we did not observe a spontaneous evolution to tertiary creep.

231 Experiments performed under conditions of pore fluid pressurization show higher
232 values of creep velocity compared to cases without injection. We measured creep
233 velocity of 40 nm/s ($\dot{\gamma} = 2 \times 10^{-4} \text{s}^{-1}$) when fluid pressure was increased at 0.2 MPa/12min
234 (Fig. 4a), and creep velocity of 50 nm/s ($\dot{\gamma} = 3 \times 10^{-4} \text{s}^{-1}$) for injection at 1 MPa/h (Fig. 4b)
235 for experiments performed at 90% of τ_{ss} . The evolution of fault slip is affected by fluid
236 injection, showing a net deviation from the curve obtained under constant P_f . For the
237 experiment at injection rate of 0.2 MPa/12min, during the early stages of injection (i.e.
238 $13 < P_f < 14$ MPa), fault slip began to slowly increase with a marked deviation from the
239 constant P_f experiment at $P_f > 14$ MPa (Fig. 4a). Similarly, the experiment performed at
240 injection of 1 MPa/h showed a net deviation from the constant P_f curve at $P_f = 14$ MPa
241 (Fig. 4b). For both injection rates, we did not observe variations in secondary creep rate
242 with increasing pore pressure. For shear stress at 80% of τ_{ss} increasing pore fluid pressure
243 caused creep acceleration, with slip velocities increasing to values of 22 nm/s ($\dot{\gamma} = 9 \times 10^{-5}$
244 s^{-1}) for injection at 0.2 MPa/12min (Fig. 5a) and 15 nm/s ($\dot{\gamma} = 8.4 \times 10^{-5} \text{s}^{-1}$) for fluid
245 injection at 1 MPa/h (Fig. 5b). Under creep loading, with constant shear stress boundary
246 conditions, fluid pressurization caused secondary creep to deviate from the hydrostatic
247 case as soon as P_f was increased.

248 The onset of tertiary creep is marked by a deviation from steady secondary creep
249 and is characterized by an acceleration of slip that spontaneously evolves into dynamic
250 failure. For experiments with creep shear stress of 90% of τ_{ss} , tertiary creep began when
251 the effective normal stress approached the failure envelope for both of our injection

252 procedures (Fig. 4). The onset of tertiary creep was observed after 200 minutes of
253 injection at 0.2 MPa/12min, at a pore fluid pressure of 16.2 MPa, after the fault
254 accumulated ~ 50 μm of slip corresponding to creep induced shear strain of $\gamma=0.07$ (Fig.
255 4a). For the experiment with injection at 1 MPa/h tertiary creep began after 180 minutes,
256 at $P_f=16$ MPa, with ~ 70 μm of slip accumulated, corresponding to $\gamma =0.08$ (Fig. 4b).
257 The shorter time to failure at injection of 1 MPa/h is in agreement with the slightly faster
258 creep velocity during secondary creep (Fig. 2). In all of our experiments, once the
259 acceleration begins fault slip increases exponentially. Slip velocity reached 2.5 mm/s
260 after 12 mm of slip, at which point we had to stop the experiment due to the finite
261 maximum displacement.

262 At 80% of τ_{ss} the time to failure is considerably longer than for the 90% τ_{ss} case.
263 The onset of tertiary creep occurred after 310 minutes for the experiment at injection of
264 0.2 MPa/12min with an accumulated slip of ~ 50 μm corresponding to $\gamma=0.03$ (Fig. 5a).
265 For the experiment with injection at 1 MPa/h the onset of tertiary creep occurred after
266 390 minutes once slip had reached ~ 47 μm , corresponding to $\gamma=0.05$ (Fig. 5b). During
267 dynamic failure fault slip velocity is characterized by peak values of ~ 3 mm/s after 12
268 mm of accumulated slip. Here again the shorter time to failure at injection of 0.2
269 MPa/12min is in agreement with the slightly faster creep velocity during secondary creep
270 (Fig. 5). In addition, under this shear stress boundary condition dynamic failure
271 propagates once the effective normal stress overcomes the Coulomb-Mohr failure
272 envelope and acceleration is more abrupt in comparison to the 90% τ_{ss} case (e.g. Fig. 4
273 vs. Fig. 5).

274 **3.3 Volumetric strain and layer thickness evolution.**

275 Tracking volume changes during deformation can reveal important details of the
276 micromechanical behavior associated with fault slip. In our experiments the changes in
277 layer thickness are a direct proxy for volume strain and fault porosity during deformation
278 (Samuelson et al., 2009). In Figure 6 (upper panels) we show the evolution of gouge layer
279 thickness, with values offset at the onset of the creep stage for comparison purposes
280 (Table 1). During the experiments performed at 90% of τ_{ss} we document a first stage
281 during which fault gouge undergoes minor compaction with an evolution to constant
282 values during secondary creep. As the pore fluid pressure is increased and the failure
283 envelope is approached, fault gouge begins to dilate, reaching a peak at the onset of
284 dynamic failure after which fault gouge undergoes abrupt compaction that persists at high
285 slip velocities (Fig. 6a). When the applied shear stress is reduced to 80% of τ_{ss} , fault
286 gouge undergoes greater compaction at the beginning of the creep test in comparison to
287 the 90% of τ_{ss} case. Compaction persists for the first stages of injection until a quasi-
288 steady state layer thickness is achieved (Fig. 6b). Dilation begins as the failure envelope
289 is approached with fault dilation that accelerates and culminates to a peak, and as
290 dynamic failure propagates, the fault abruptly compacts.

291 It is important to note that the evolution of layer thickness is particularly sensitive
292 to initial starting condition such as grain packing, porosity and the degree of shear
293 localization. The variability that we observe in our experiments is expected, based on the
294 minor variations in initial porosity and grain packing from sample to sample (Table 1).
295 Several suites of trial experiments and reproducibility tests show that even with extreme
296 attention to detail during sample preparation, and following the same experimental
297 protocol, it is impossible to control the evolution of gouge deformation during the first

298 stages of deformation (i.e. constant strain rate and hold period). During these stages,
299 variations in shear localization affect fault gouge porosity. However, the striking
300 similarities that we observe in the evolution of gouge layer thickness across multiple
301 experiments, even if the absolute values are slightly different, makes us confident in the
302 voracity of our observations.

303 **3.4 Hydrological behavior.**

304 Diffusivity and flow of fluid within fault gouge during shear is an important
305 parameter that can influence fault slip behavior. Fluid pressure controls the stress state
306 along with porosity of the fault gouge (Segall and Rice, 1995; Wibberley, 2002; Faulkner
307 et al., 2010). In Figure 6 (lower panels), we show the evolution of the up- and down-
308 stream pore fluid pressure during experiments using both of our injection procedures. For
309 injection at 1 MPa/hr, in response to the instantaneous increase in the up-stream fluid
310 pressure, the fluid pressure front rapidly diffuses within the fault and it equilibrates at the
311 down-stream intensifier with an average time lag of 391 seconds and 328 seconds for the
312 90% and 80% of τ_{ss} experiments respectively. For the case of injection at 0.2 MPa/12min
313 the time lag for equilibration is on average 130 seconds for the experiments at 90% of τ_{ss}
314 and 133 seconds for experiments at 80% of τ_{ss} . Given the relatively high permeability of
315 the fault gouge (i.e. $\sim 10^{-17} \text{m}^2$ and Fig. 3b), the observed time lag is short in comparison
316 with the total time of fluid pressure rise, with transient pressure representing $\sim 9\%$ of the
317 total step time at 1 MPa/h and $\sim 18\%$ for injection at 0.2 MPa/12min. We also note that
318 the values for equilibration do not show any systematic trend with increasing pore fluid
319 pressure and they are not associated with fault dilation/compaction.

320 **3.5 Microstructural observations.**

321 At the end of selected experiments, we collected the fault zones for Scanning
322 Electron Microscopy (SEM) analysis (Fig. 7). Shear is accommodated by grain size
323 reduction and cataclasis in the gouge, where clasts of dimension comparable with the
324 starting material are highly fractured and finer grains are angular (Fig. 7b). Deformation
325 is localized along R1-planes (Logan, 1979) and sharp B-planes with a thickness of ~10-
326 20 μm where we observe intense grain size reduction with nanograins surrounded by a
327 very fine matrix (Fig. 7 b, d). Evidence of pressure solution is visible within the B-planes
328 in the form of grain-to-grain indentation (i) (Fig 7d). On the surface of bigger grains we
329 also observe dissolution pits as a further indication of rock fluid interaction (Fig. 7c). Our
330 microstructural observations are consistent with previous works showing a similar fault
331 zone structure for carbonate bearing fault gouge sheared at a range of slip velocities and
332 stress boundary conditions (e.g. Verberne et al., 2015; Carpenter et al., 2016).

333 **4. Discussion**

334 **4.1 Mechanics of fault gouge creep**

335 We investigated the conditions that lead to dynamic slip instability during fault
336 zone fluid pressurization. In creep experiments the fault zone deformation history can be
337 divided into three main stages that are persistent at the different applied shear stresses and
338 injection procedures, but with different absolute values (Fig. 8). The first stage is
339 associated with primary creep and begins at the onset of the creep test. This stage is
340 characterized by fault zone compaction and a deceleration in slip velocity. We find a
341 positive relation between the amount of compaction and the applied shear stress, where
342 for experiments at 90% of τ_{ss} we observe less compaction than at 80% of τ_{ss} . During this
343 stage micro-crack closure, changes in grain packing and contact processes such as

344 pressure solution produce fault zone compaction (Fig. 7). The larger amount of
345 compaction for the 80% τ_{ss} case implies that the fault zone undergoes greater
346 strengthening with lesser shear driven dilation due to creep consolidation facilitated by
347 asperity contact growth and interparticle slip within the localized shear zones. At the end
348 of stage one, the fault zone reaches a steady state porosity without further compaction.
349 Stage two begins during the first phases of fault zone pressurization, corresponding with
350 secondary creep, during which the fault slips at steady state porosity (Fig. 6a, b and 8).
351 We find that secondary creep rates increase with the applied shear stress, with the
352 experiments at 90% of τ_{ss} showing higher creep rates than at 80% of τ_{ss} by a factor of ~ 2 .
353 This behaviour is in agreement with numerous creep studies on intact rocks (e.g. Kranz
354 and Scholz, 1977; Baud et al., 1997; Heap et al., 2009; Brantut et al., 2013). As fluid
355 pressure is further increased and the stress state approaches the failure envelope, fault
356 zone dilation begins during stage three. In general, we find that the fault begins to dilate
357 at values of $P_f=15$ MPa for creep at 90% of τ_{ss} and $P_f=16$ MPa for creep at 80% of τ_{ss} ,
358 which correspond to effective stresses below the Coulomb failure envelope. During this
359 stage fault creep is still steady. A further increase in pore fluid pressure causes the fault to
360 meet the stress state for reactivation and we observe different slip evolution depending on
361 the applied shear stress. For 90% of τ_{ss} (Fig. 8a and c), this stress state marks the onset of
362 tertiary creep, during which fault zone dilation increases log-linearly with slip velocity
363 until a critical point, slip velocity of ~ 0.3 mm/s, where the fault abruptly compacts and
364 fails dynamically, with slip velocity >1 mm/s (stage four). For 80% of τ_{ss} (Fig. 8b and d),
365 as the stress state reaches the failure criterion the fault zone begins to accelerate and
366 dilation evolves log-linearly with slip velocity. However, in this case, fault gouge failure

367 is achieved via a further increase in pore fluid corresponding to a stress state beyond the
368 failure criteria (Fig. 8d). This indicates that that the fault gouge acquired cohesion during
369 the previous stages due to longer fault creep, 350 vs. 150 minutes, with a more efficient
370 fluid-rock interaction resulting in a larger healing and cementation. This observation is
371 well coupled with the evolution from a log-linear behavior to a power law type evolution
372 of gouge dilation during acceleration, such that in order to overcome the interparticle
373 cohesion more dilation is required. Physico-chemical processes such as interparticle
374 pressure solution, as observed within the localized zones of the fault gouge (Fig. 7), can
375 increase the contact area (either the quality and/or the quantity) at particle junctions
376 resulting in an overall strengthening of the fault gouge, which is in good agreement with
377 our interpretation (e.g. Bos and Spiers, 2002). The peak in dilation marks the onset of
378 dynamic slip and fault gouge compacts at velocities $>1\text{mm/s}$ (stage four).

379 **4.2 Rate- and State- Friction vs. dynamic slip of pressurized fault gouge**

380 The steady state rate dependence of friction for calcite fault gouge at the stresses
381 and fluid pressures of our study (i.e. $\lambda=0.5$) (Fig. 4c and 5c) is clearly velocity
382 strengthening, which should produce intrinsically aseismic creep (Fig. 3 in Scuderi and
383 Collettini, 2016). In addition, the criterion for fault frictional stability described in
384 Equation 4 predicts that an increase in fluid pressure should tend to stabilize fault slip,
385 because it reduces the critical rheological stiffness. However, fluid pressurization during
386 our creep experiments causes accelerated fault creep that evolves in dynamic slip
387 instability at values of λ characteristic of a velocity strengthening behavior. In this
388 context, we face a contrasting effect of the influence of fluid pressure on fault slip
389 stability when evaluated with a RSF or with a creep approach. Here, we posit that fault

390 weakening induced by fluid pressurization overcomes the second order rate strengthening
 391 effect, resulting in fast acceleration and dynamic slip.

392 To illuminate the details of the relationship between friction rate dependence and
 393 effective normal stress, we evaluate the interaction between fault zone deformation and
 394 applied stress field following the early work of *Frank* [1965] and many others (e.g.
 395 Marone et al., 1990, Beeler and Tullis, 1997; Bos and Spiers, 2002; Niemeijer et al.,
 396 2008). Considering a closed system that obeys the first law of thermodynamics, it is
 397 possible to express the energy balance for a representative unit volume of fault gouge
 398 during deformation as (Bos and Spiers, 2002):

$$399 \quad \tau \dot{\gamma} + (\sigma_n - P_f) \dot{\epsilon} = \int_V \sum_m \dot{\Delta}_m dV \quad (6)$$

400 where τ is the shear stress, $\dot{\gamma}$ is the shear strain rate, $(\sigma_n - P_f)$ represents the effective
 401 normal stress (compression positive), $\dot{\epsilon}$ is a compactional strain rate (compaction
 402 negative), V is the volume and $\dot{\Delta}_m$ represents a specific dissipation rate by process m . In
 403 this context, the right-hand side of Equation 6 represents the sum of all microscale
 404 dissipative processes per unit volume that include grain fracture, dilatancy, frictional
 405 sliding of grain contacts, pressure solution and crystal plasticity. Rearranging Equation 6
 406 in terms of shear stress yields:

$$407 \quad \tau = \tau_x + \frac{d\epsilon}{d\gamma} (\sigma_n - P_f) \quad (7)$$

408 where τ_x represents the contribution to shear strength of all energy dissipative processes
 409 operating in the gouge and it is expressed as:

$$410 \quad \tau_x = \int_V \sum_m \frac{\dot{\Delta}_m}{d\gamma} dV \quad (8)$$

411 For a thin gouge layer, such as in our case, and for our experimental geometry we can
412 express $d\varepsilon = dV/V$ and $d\gamma = d\delta/h$ where V is sample volume, δ is the fault slip and h is the
413 layer thickness. The volume strain can be expressed as $d\varepsilon = dhA/Ah$, where A is the
414 nominal frictional contact area and thus the ratio $d\varepsilon/d\gamma$ reduces to $dh/d\delta$ which are all
415 measurable quantity in our experiments (Fig. 9) (Marone et al., 1990; Beeler and Tullis,
416 1997). During our creep experiments we impose a constant shear stress on the fault gouge
417 so that the sum of the changes in the micro-mechanical processes at grain-to-grain
418 contacts, τ_x , and changes in the effective normal stress times fault dilation/compaction
419 with slip, $dh/d\delta$, has to remain constant. During the initial stage of fault creep, when fluid
420 pressure is nearly constant (stage 2, blue paths in Fig. 8), we observe fault compaction
421 indicating that during this stage τ_x increases as a result of physico-chemical processes at
422 grain contacts (Fig. 7). With increasing fluid pressure (i.e. effective stress decrease), the
423 energy of the system is unbalanced, and to maintain the system at equilibrium (i.e.
424 constant shear stress) the fault zone has to dissipate the energy by dilating. Assuming that
425 grains slide over each other (i.e. no grain rolling) with increasing dilation the fault zone
426 begins to accelerate: note that in our data the onset of dilation always precedes the onset
427 of tertiary creep, e.g. Fig. 8 and 9b, in agreement with previous experimental work and
428 models (Chen and Spiers, 2016). At the critical stress state for reactivation, fault gouge
429 reaches a maximum attainable value of dilation (i.e. Fig. 9c), beyond which the fault
430 cannot dilate further. At this point, stage 4, fault dilation is no longer an efficient
431 mechanism for energy dissipation and the fault system reacts with fracturing and shear
432 localization resulting in dynamic slip propagation (Fig. 9). Fracturing and shear

433 localization are significant energy dissipative processes that increases τ_x in agreement
434 with the observed compaction during stage 4 (Fig. 9a and b).

435 The duality between the rate strengthening behavior retrieved from RSF analysis
436 and the observed nucleation of dynamic instability can be explained by considering that
437 RSF parameters are evaluated at a steady state frictional sliding regime and at imposed
438 slip velocity. This implies that during steady state shear the system is in a dynamic
439 equilibrium with the ongoing time dependent compaction balanced by slip dependent
440 dilation. If slip velocity increases at constant effective normal stress, for either high or
441 low fluid pressure, the frictional response will result from the evolution of the asperity
442 contact population in the local stress field. This implies that the standard model for
443 earthquake nucleation, based on RSF constitutive parameters needs to be modified to
444 account for the effect of fluid overpressure to accurately predict the seismic risk
445 associated with fluid injection.

446 **4.3 Implication for induced seismicity**

447 In the context of human-induced seismicity, understanding the physical
448 mechanisms that lead faults to slip seismically or aseismically in response to pressurized
449 fluids is a primary goal to mitigate the seismic risk during injection. The potential to
450 nucleate a seismic instability, as evaluated in dynamic nucleation models based on RSF
451 principles, requires an initial fault zone rheology characterized by a velocity weakening
452 behavior (e.g. Urpi et al., 2016). However, under upper crustal boundary conditions (i.e.
453 depth \sim 6-7 km) and for temperatures $<120^\circ\text{C}$, laboratory experiments have shown that a
454 great number of fault gouges with characteristic lithologies observed or inferred to host
455 induced earthquakes (i.e. carbonates, shales and granites), show predominantly velocity

456 strengthening behavior, intrinsic of aseismic creep. These results contrast with
457 observations of induced earthquakes during wastewater injection where seismicity is
458 generally confined to the upper 6-7 km and it is generally related with peaks in pore fluid
459 injection rates at the well head (e.g. Improta, 2015; Yeck et al., 2017). Large scale field
460 experiments have also shown that pressurized fluids reactivate faults where complex
461 seismic behaviors are observed (Guglielmi et al., 2015). Our results show that even for
462 small changes in fluid pressure the effect of normal stress on fault strength and stability
463 outweighs the rate and state dependent effects promoting fault unstable behavior.

464 **5. Conclusion**

465 Our experiments shed light on the physical processes responsible for fluid
466 induced fault deformation. We show that in a laboratory fault, dynamic slip instabilities
467 can be induced by an increase in pore fluid pressure once the critical stress state for
468 reactivation is met, even if the fault is characterized by velocity strengthening frictional
469 behavior. Under these conditions the instability is driven by an energy unbalance caused
470 by a decrease in effective normal stress and fault zone weakening. Under a broad range of
471 conditions, this effect outweighs the impact of the second order rate and state effects on
472 fault zone frictional strength. We posit that to mitigate the risk of induced seismicity a
473 careful characterization of the stress field surrounding the fault where fluid will be
474 injected it is essential and fluid pressure should be maintained below the critical stress
475 state for reactivation.

476

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484 should be addressed to marco.scuderi@uniroma1.it. All the data are available via FTP
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486 **Author Contribution:** All the authors contributed to the experimental design, data
487 interpretation and writing. M.M. Scuderi conducted the experiments and performed data
488 analysis.

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490

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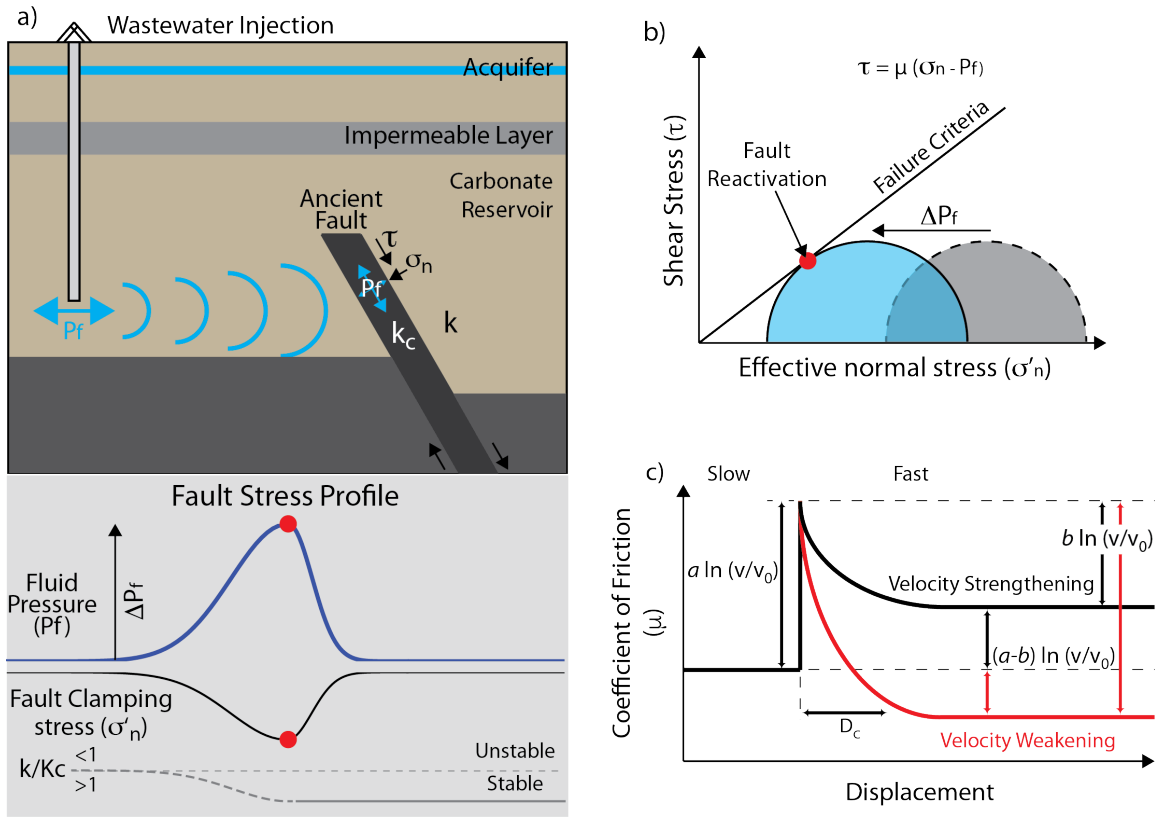
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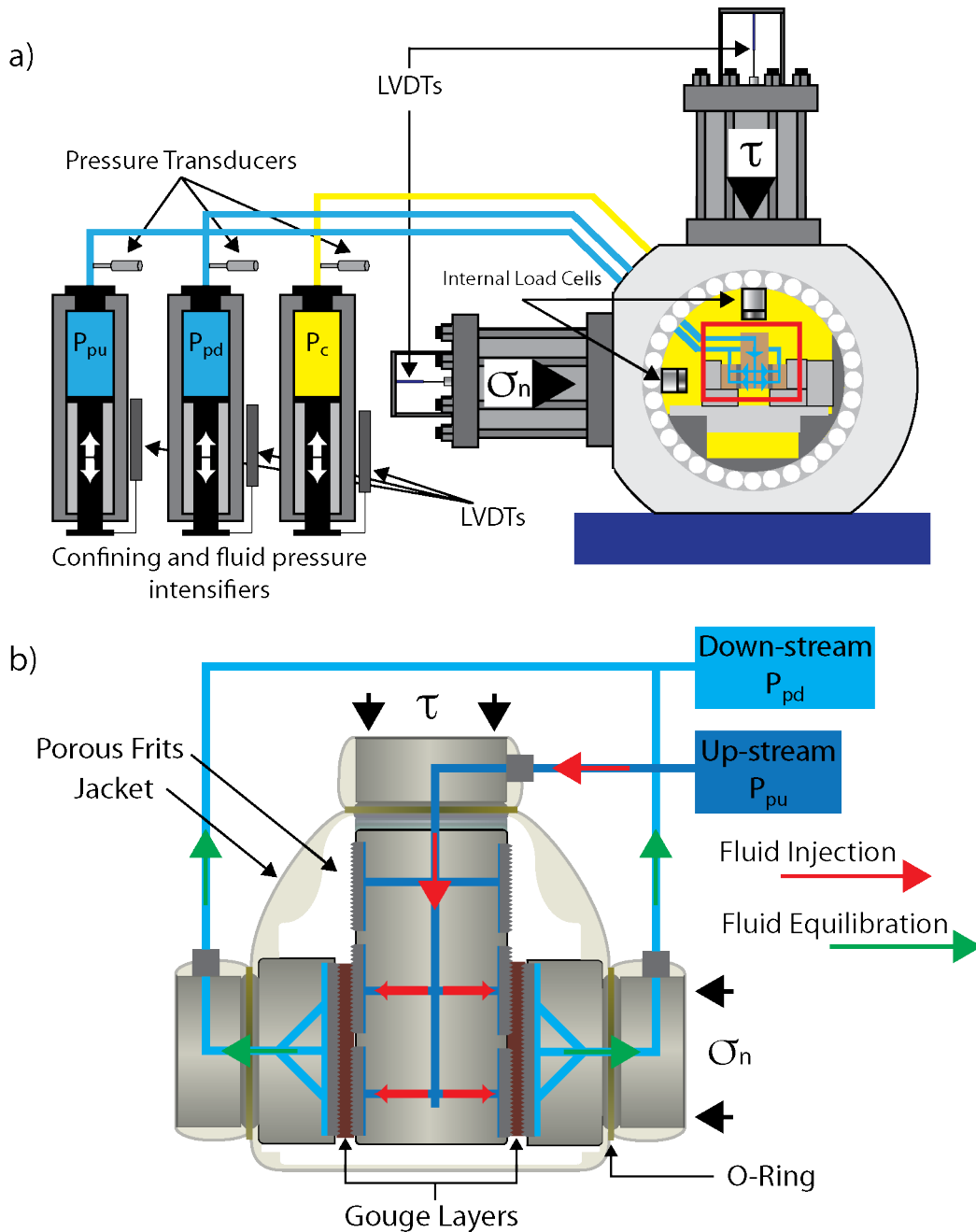
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630 **Figures**



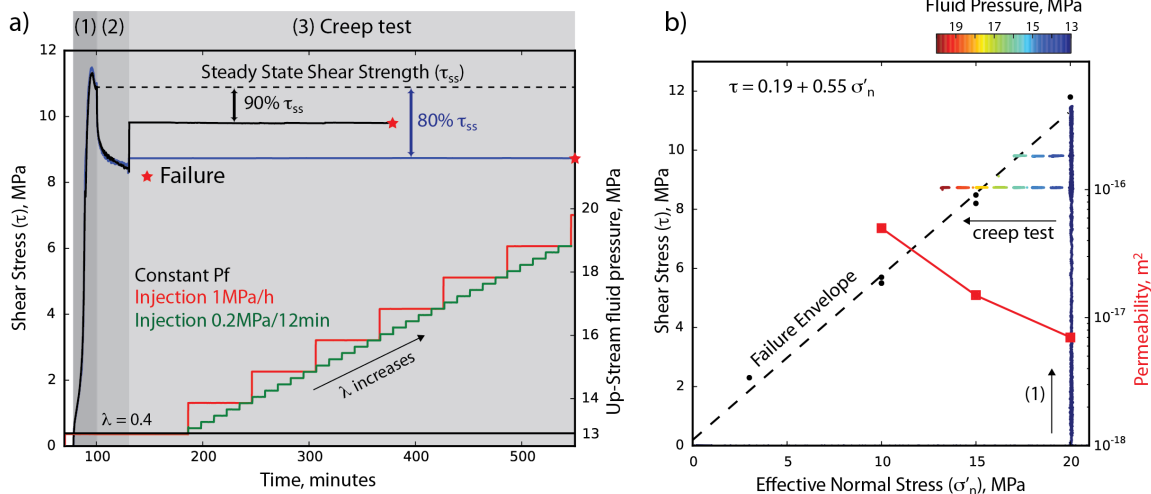
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632 **Figure 1.** Schematic illustrations of (a) mechanism(s) for induced seismicity associated
 633 with fluid injection and (lower panel) the stress state around an injection well. In
 634 response to fluid injection, the fluid pressure front diffuses and modifies the stress field
 635 around faults, causing fault reactivation. (b) Coulomb-Mohr diagram for shear failure and
 636 (c) the principles of rate- and state- friction (RSF). When the initial stress state of a fault,
 637 gray circle in (b), is perturbed by an increase in fluid pressure (ΔP_f), the conditions for
 638 fault reactivation are favored, blue circle in (b). Under these conditions, the fault
 639 frictional stability is evaluated via the RSF behavior (c). An increase in sliding velocity
 640 causes an instantaneous increase of the frictional strength that evolves in two main
 641 fashions. If the frictional strength increases the fault has the characteristic “velocity
 642 strengthening” behavior which leads to stable sliding (black line). Whereas, in the
 643 “velocity weakening” regime increased slip velocity causes a decrease in frictional
 644 strength, and the fault has the potential to nucleate a seismic instability (red line).
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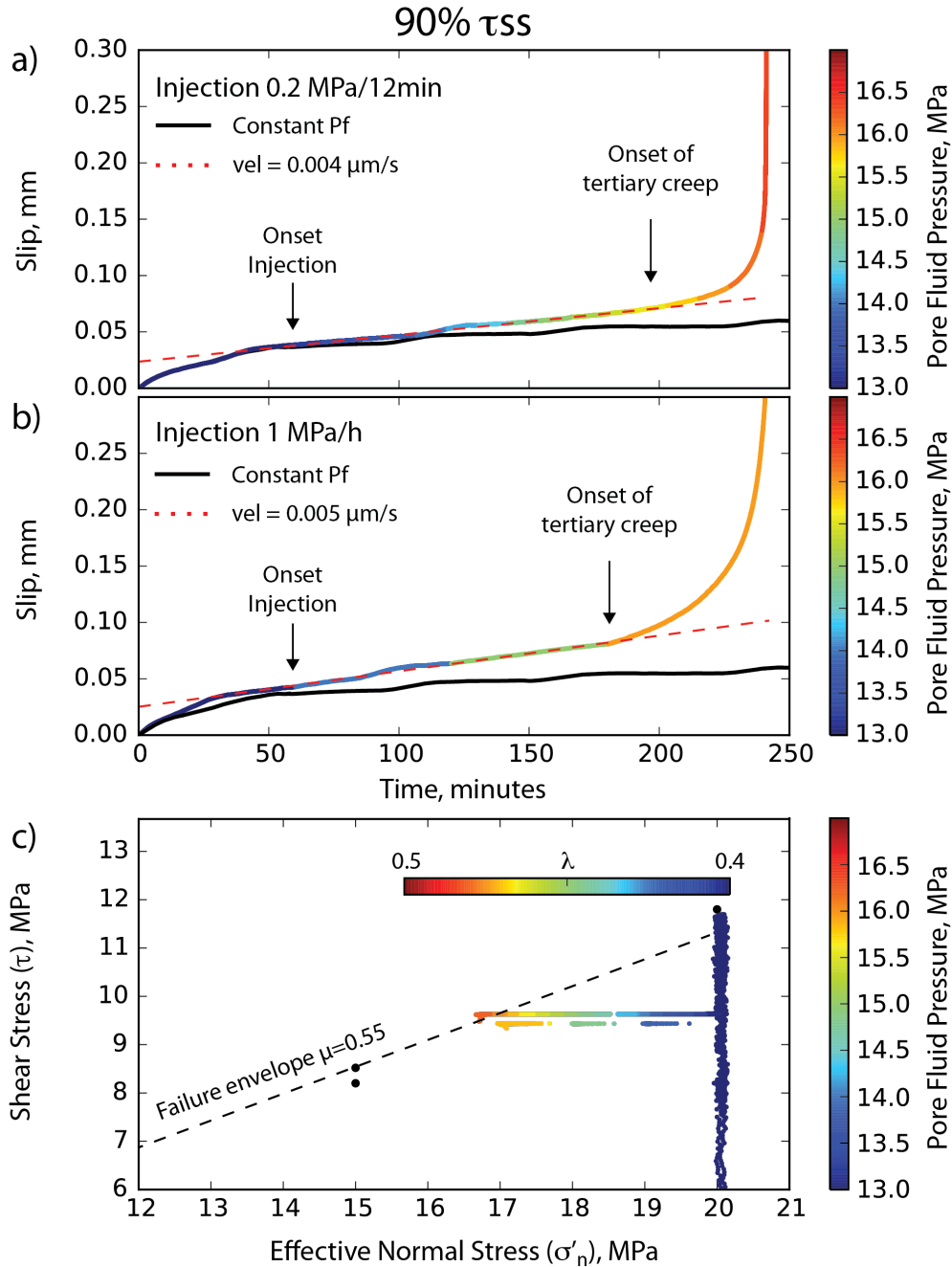
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647 **Figure 2.** Schematic of the experimental configuration. (a) BRAVA (Brittle Rock
 648 deformAtion Versatile Apparatus) deformation machine showing the double direct shear
 649 configuration (red box) within a pressure vessel. Three intensifiers are used to pressurize
 650 pore fluid within the experimental fault gouge (P_{pu} and P_{pd}) and to apply confining
 651 pressure (P_c). (b) Details of the sample assembly in the double direct shear configuration.
 652 During the experiments we increase fluid pressure from the up-stream reservoir (red
 653 arrow) and record fluid pressure at equilibrium at the down-stream reservoir (green
 654 arrow) after the fluid pressure front diffuses within the gouge layers.
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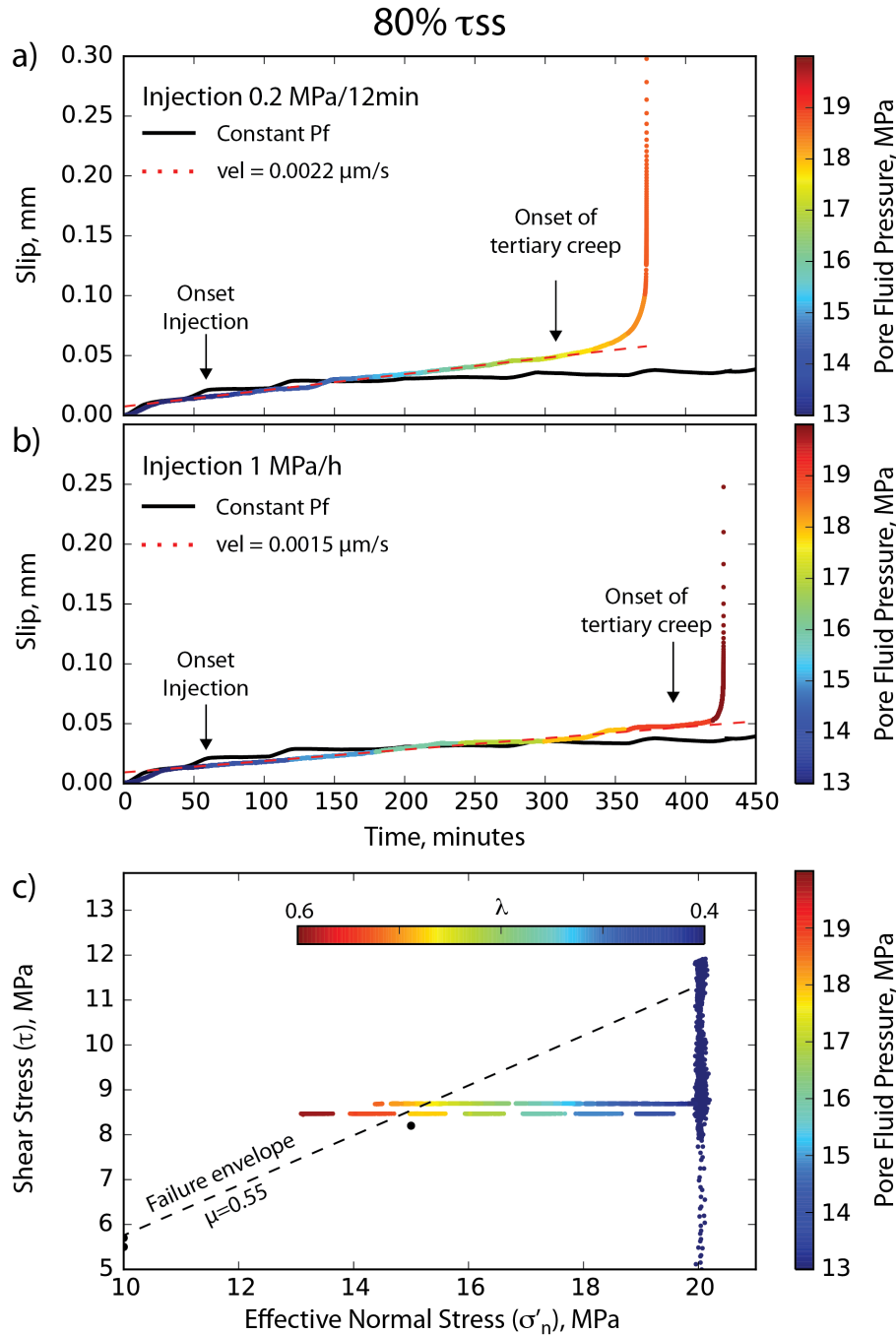
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Figure 3. Experimental Procedure. (a) Typical experimental curves for two experiments (exp. num. b583 in black and b589 in blue) that show the evolution of shear stress as a function of time. After the first stage at constant displacement rate (1) the fault relaxes (2) and then we fix a constant shear stress at either 80% (blue curve) or 90% (black curve) of the steady state shear strength (3). During the creep tests, we increase pore fluid pressure (bottom curves) at either 1MPa/h (red curve) or 0.2 MPa/12 minutes (green curve) and monitor the resulting fault slip. For reference, we also performed experiments at constant pore fluid pressure (black line). (b) Coulomb failure diagram where we report the experimental data shown in (a) along with the permeability measured at different σ'_n .



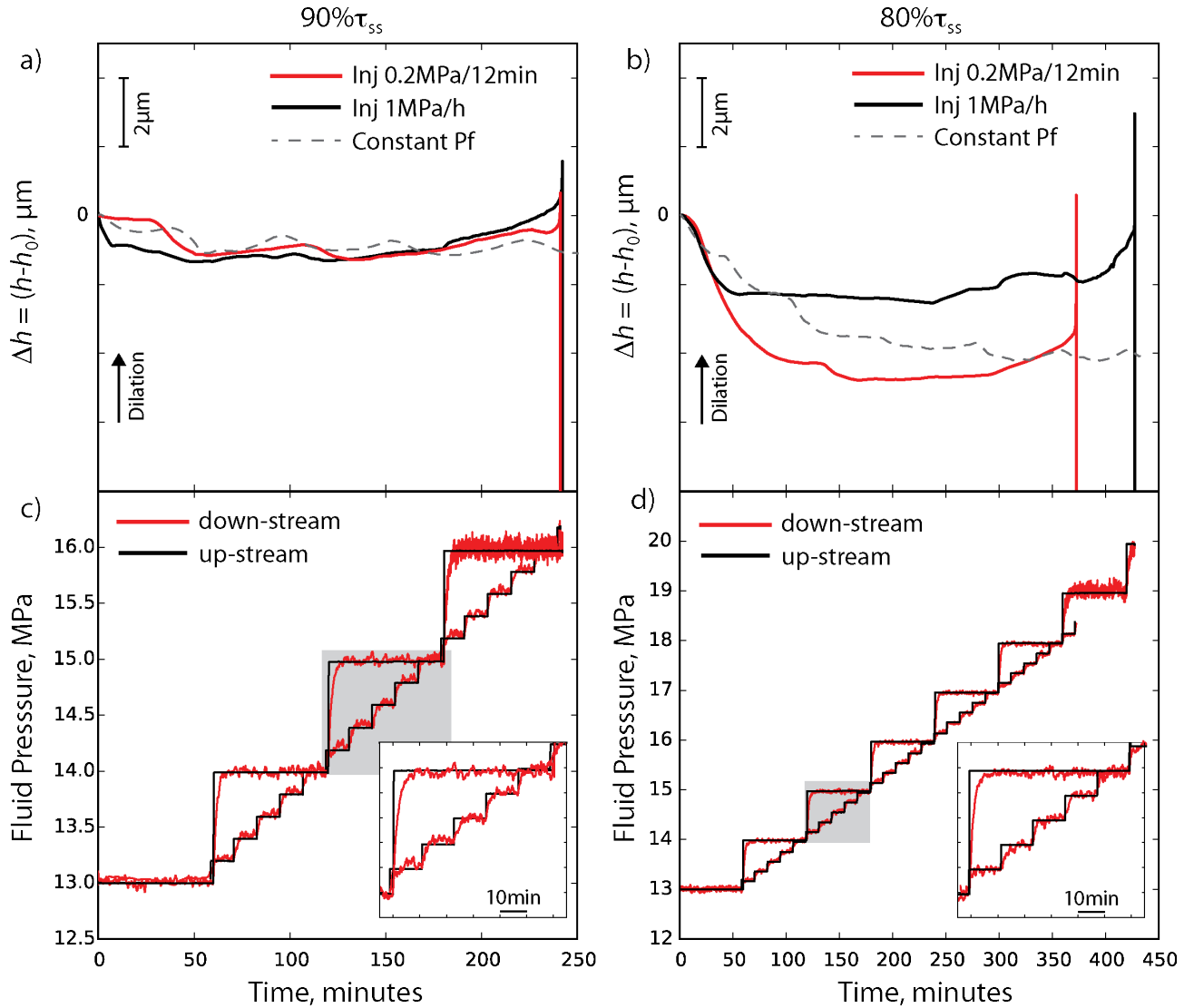
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Figure 4. Raw data showing the evolution of fault slip for the creep experiments performed at 90% of τ_{ss} under fluid injection conditions of (a) 0.2 MPa/12minutes and (b) 1MPa/h (exp. num. b593 and b595 respectively). In black we report the creep curve at constant Pf for reference (exp. num. b590). (c) Coulomb failure diagram showing the corresponding stress path for the curves shown in (a) and (b) in relation to the failure envelope. Values of the pore fluid factor, λ , are also reported. Note that the stress path during creep for the experiment at 1 MPa/h has been offset by 0.1 MPa to avoid overlap with the stress path at 0.2 MPa/12min.



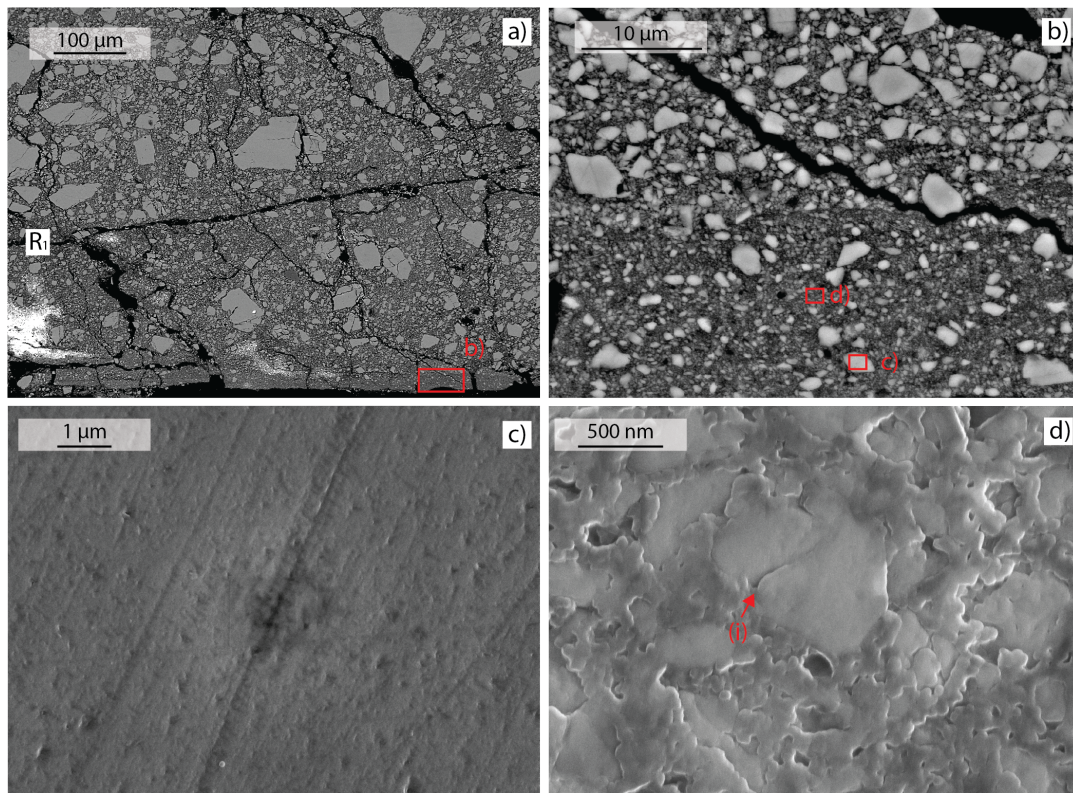
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 678 **Figure 5.** Raw data showing the evolution of fault slip for the creep experiments
 679 performed at 80% of τ_{ss} under fluid injection conditions of (a) 0.2 MPa/12minutes and (b)
 680 1MPa/h (exp. num. b592 and b644 respectively). In black we report the creep curve at
 681 constant Pf for reference (exp. num. b594). (c) Coulomb failure diagram showing the
 682 corresponding stress path for the curves shown in (a) and (b) in relation to the failure
 683 envelope. Values of the pore fluid factor, λ , are also reported. For injection at 0.2
 684 MPa/12min fault failure propagates at $P_f=18.4$ MPa, whereas for the injection at 1 MPa/h
 685 at $P_f=20$ MPa, corresponding to a stress surplus of $\sigma'_n=0.7$ MPa (0.2 MPa/12min) and σ'_n

686 =2.2 MPa (1 MPa/hr). Note that the stress path during creep for the experiment at
687 1MPa/h have been offset of 0.1 MPa to avoid overlap with the stress path at 0.2
688 MPa/12min.
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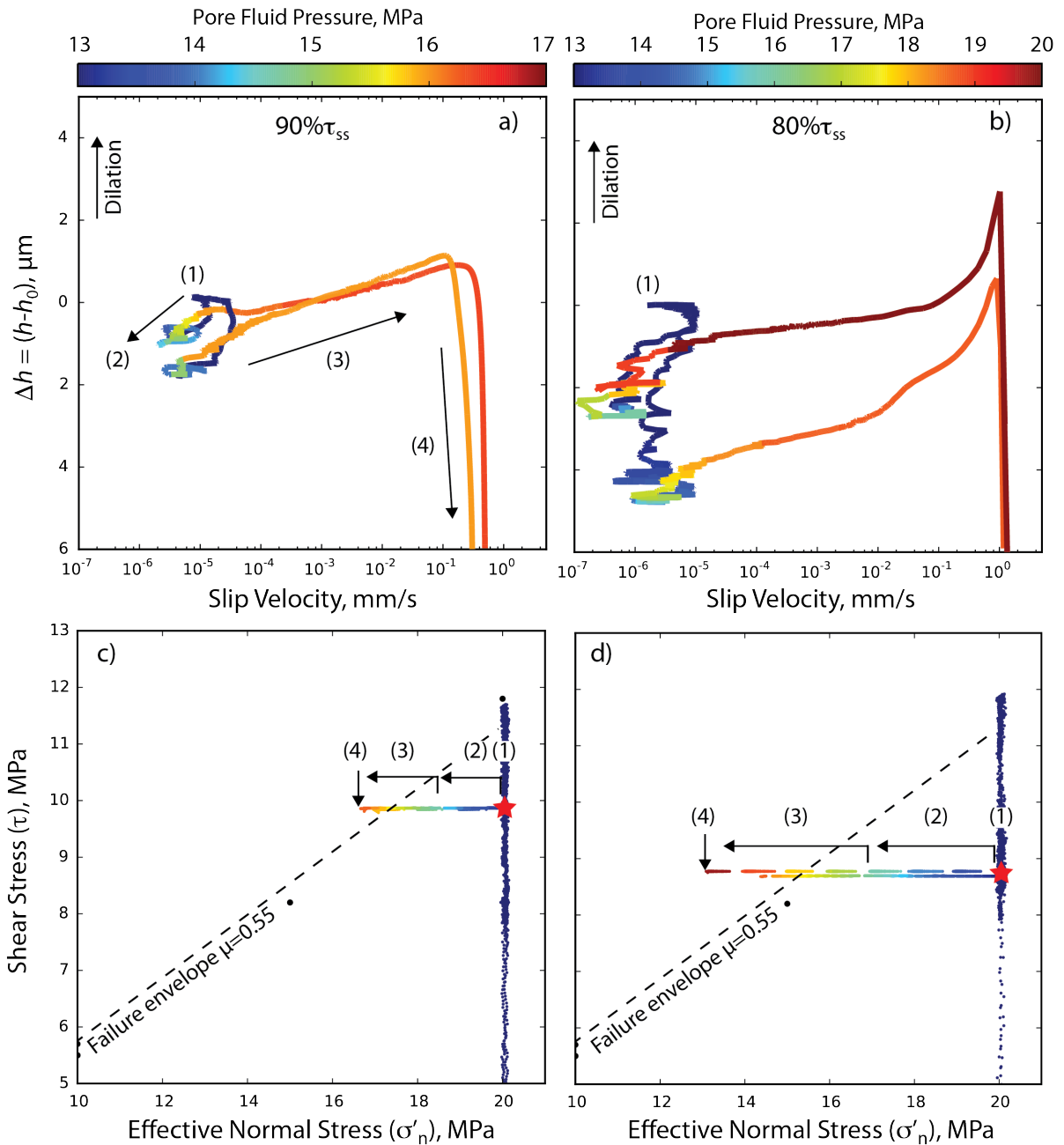


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Figure 6. Volumetric strain and injection curves. Top panels show the evolution of the changes in layer thickness during creep experiments performed at (a) 90% of τ_{ss} and (b) 80% of τ_{ss} . Bottom panels: fluid injection curves showing the equilibration of fluid pressure between the up-stream reservoir (injection side in black) and the down-stream reservoir (in red) after passing through the gouge layers for the (c) 90% of τ_{ss} and (d) 80% of τ_{ss} cases and both the injection procedures (i.e. 1MPa/h and 0.2 MPa/12min). Insets in (c) and (d) show details for the pressure steps in the gray box.

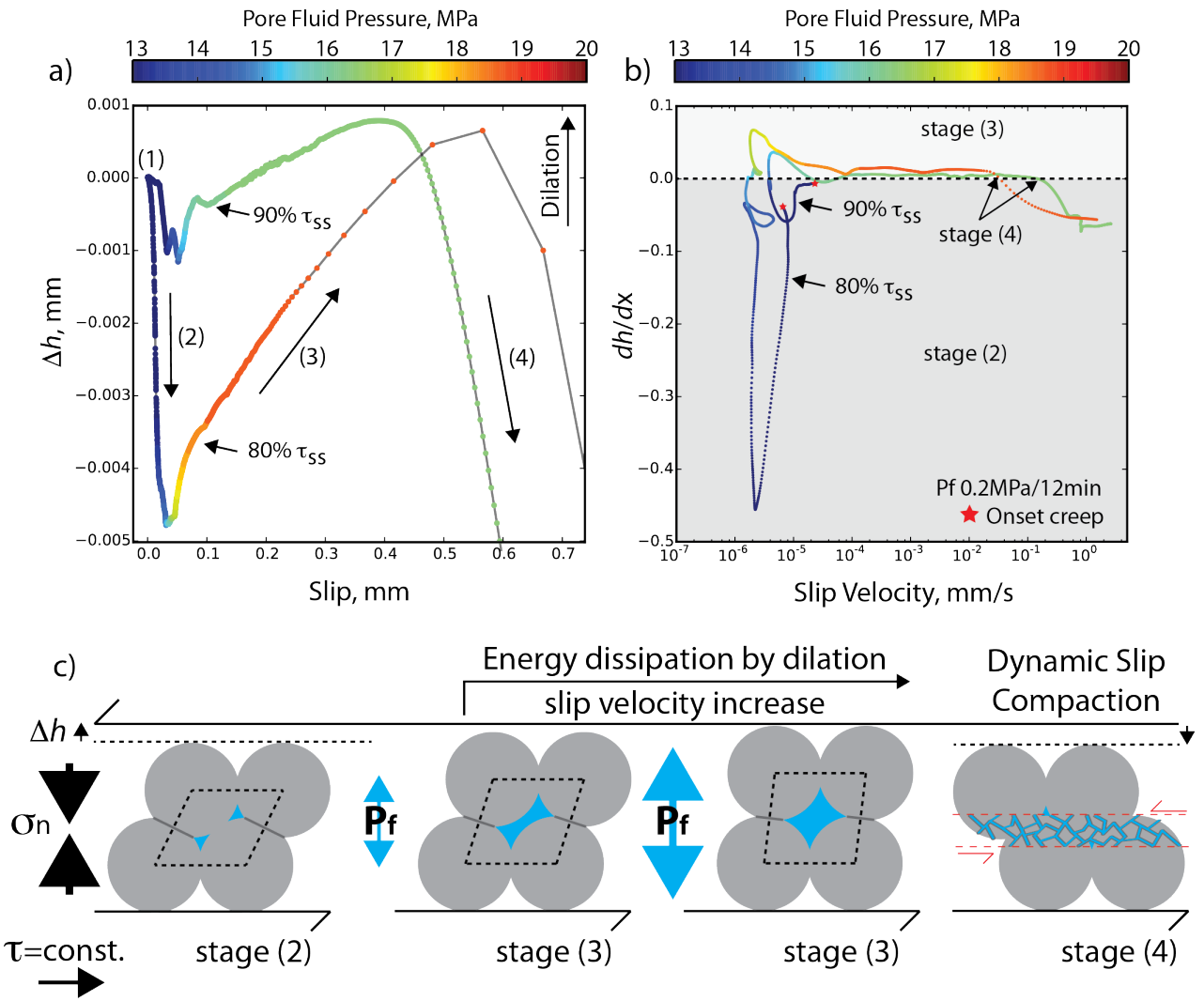


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700 **Figure 7.** Fault zone microstructure recovered after shear for a representative case at 80%
701 of τ_{ss} (exp. num. b592). (a) Shear localizes along R1 planes and sharp B-planes at the
702 layer boundary. (b) Zoom on the B-plane showing strong grain size reduction and larger,
703 rounded clasts. (c) Dissolution pits on the surface of a bigger grain in the localized slip
704 zone. (d) Details of physicochemical processes within the B-planes with grain indentation
705 (i) and cemented nanoparticles indicating that pressure solution is most likely acting
706 during fault creep.
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Figure 8. Evolution of layer thickness as a function of slip velocity for experiments performed at (a) 90% of τ_{ss} and (b) 80% of τ_{ss} at both the injection rate of 1 MPa/h and 0.2 MPa/12min. We observe three main stages for fault deformation (see text for more details) that correspond to different stress states as shown in the corresponding Coulomb failure diagrams (c and d).



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Figure 9. Conceptual model for energy unbalance and dynamic slip. Top panels: (a) evolution of gouge layer thickness as a function of slip for experiments performed at 80% and 90% of τ_{SS} at injection rate of 0.2 MPa/12min. (b) Corresponding evolution of fault gouge deformation (dh/dx in equation 6) as a function of slip velocity highlighting the different stages of fault deformation. (c) Conceptual model describing the evolution of fault zone deformation associated with different stages of shear, based on the mechanical data in (a) and (b).

Fault strength and permeability										
Exp	ρ g/cm ³	σ_n MPa	P_c MPa	P_f MPa	σ'_n MPa	τ_{ss} MPa	γ Measure Perm.	Permeability m ²		
b581	1.345	2	15	7	10	5.55	6.6	5e-17		
		10	15	10	15	8.21	7.8	1.5e-17		
		18	15	13	20	11.85	10.2	7e-18		
b582	1.367	2	15	7	10	5.53	6.8	4.5e-17		
		10	15	10	15	8.18	8.1	1e-17		
		18	15	13	20	11.79	10.7	6e-18		
Creep Experiments										
Exp	ρ g/cm ³	μ_{peak}	μ_{ss}	τ_{ss} MPa	δ onset creep mm	γ onset creep	LT onset shear mm	LT Onset Creep mm	$\tau\%$ relative to τ_{ss}	Injection procedure
b591	1.370	0.58	0.57	11.4	14.27	8.3	1.806	1.673	90%	Constant Pf
b590	1.325	0.57	0.55	11.0	13.62	8.8	1.613	1.522	90%	Constant Pf
b593	1.304	0.57	0.53	10.7	13.33	9.1	1.546	1.391	90%	0.2MPa/12min
b583	1.347	0.56	0.54	10.8	13.29	7.4	1.865	1.785	90%	1MPa/h
b595	1.337	0.58	0.57	11.4	13.93	10.3	1.438	1.308	90%	1MPa/h
b594	1.297	0.58	0.57	11.3	13.53	8.8	1.624	1.495	80%	Constant Pf
b592	1.335	0.57	0.54	10.9	14.45	8.4	1.778	1.722	80%	0.2MPa/12min
b589	1.223	0.57	0.55	11.0	12.92	9.3	1.452	1.337	80%	1MPa/h
b644	1.312	0.57	0.55	11.1	13.56	8.8	1.573	1.412	80%	1MPa/h

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Table 1. Summary of experiments and boundary conditions. Top panel: experiments performed to evaluate fault strength and permeability. We report experiment number (exp.), initial sample density (ρ), normal stress (σ_n), confining pressure (P_c), pore fluid pressure (P_f) with the resulting effective stress (σ'_n), shear stress at steady state (τ_{ss}), the shear strain (γ) correspondent to the permeability measurement. In the bottom table are reported the creep experiments. All experiments were performed at the same stress field given by: $\sigma_n=2$ MPa, $P_c=19$ MPa, $P_f=13$ MPa resulting in $\sigma'_n = 20$ MPa. Indicated are experiment number (exp.), initial sample density (ρ), peak (μ_{peak}) and steady state (μ_{ss}) coefficient of friction, with the correspondent steady state shear stress (τ_{ss}). We also

735 indicate the shear strain (γ) at the onset of the creep stage along with the absolute values
736 of layer thickness (LT) at the onset of shear and at the onset of fault creep.