A BDF2 Energy-Stable Scheme for a General Tensor-Based Model of Liquid Crystals

Guanghua Ji

Laboratory of Mathematics and Complex Systems, Ministry of Education and School of Mathematical Sciences, Beijing Normal University, Beijing 100875, China.

Received 29 October 2018; Accepted (in revised version) 7 June 2019.

Abstract. Following the scalar auxiliary variable strategy, a linear semi-discrete scheme in time for the hydrodynamic $Q$-tensor model of liquid crystal polymers is developed. It is shown that the scheme is unconditionally energy stable and uniquely solvable. Numerical simulations show the decreasing energy and the second-order convergence.

AMS subject classifications: 49Mxx, 65M12, 81T80

Key words: $Q$-tensor model, unconditional energy stability, linear second order BDF2 scheme.

1. Introduction

Liquid crystals represent an intermediate state of matter between crystalline solids and isotropic fluids. Nematic liquid crystals usually have molecular orientational order but not a positional order. The most popular mathematical model for the flow of low molecular weight nematic liquid crystals is the Ericksen-Leslie model [10], where the orientation of molecules is expressed by a unit vector $d \in S^2$. The distortional elasticity is described by the Oseen-Frank energy but in this case, only uniaxial liquid crystals can be modeled. If the orientational symmetry is broken, the Ericksen-Leslie theory fails to capture the asymmetrical feature of the system. Moreover, if defects emerge, the director model is singular and $d$ cannot be determined. In order to model liquid crystal droplets, Diegel et al. [3] coupled the Ericksen’s model for nematic liquid crystals and the Cahn-Hilliard interfacial energy equation.

An alternative method to describe the orientation of nematic liquid crystal systems consists in using a $Q$-tensor — i.e. a second-order tensor of trace zero. The reflective symmetry of the system and biaxiality are naturally built-in into tensor-based theories and defects can also be captured. Furthermore, the director model can be derived from $Q$-tensor theory for weak flows and weak elastic limits [20]. Therefore, the $Q$-tensor based hydrodynamic model is commonly used in nematic liquid crystal flows [1,5,6,18–20].

*Corresponding author. Email address: ghji@bnu.edu.cn (G. Ji)
In this work, we describe the average orientation of nematic liquid crystals with \( Q \). Set
\[
\Lambda := \{ Q \in \mathbb{R}^{3 \times 3}, \text{tr}(Q) = 0, Q = Q^T \}
\]
and consider the general Landau-De Gennes free energy functional
\[
E(Q) = \int_{\Omega} \left( \frac{K}{2} |\nabla Q|^2 + F_B(Q) \right) \, dx,
\]  
(1.1)
where the first term in the integral is the elastic energy, \( K \) a material-dependent elastic constant, \( F_B(Q) \) the bulk free energy density,
\[
F_B(Q) := \frac{\alpha}{2} \text{tr}(Q^2) + \frac{\beta}{3} \text{tr}(Q^3) + \frac{\gamma}{4} \text{tr}^2(Q^2)
\]
and \( \alpha, \beta \) and \( \gamma > 0 \) are material-dependent and temperature-dependent constants — cf. [7].

According to [1, 20, 23], the non-dimensional governing equations of nematic liquid crystal flows with hydrodynamics have the form
\[
\begin{aligned}
u_t + u \cdot \nabla u &= -\nabla p + \eta \nabla^2 u + \nabla \cdot \sigma(Q, G) - G \nabla Q, \\
\nabla \cdot u &= 0, \\
Q_t + u \cdot \nabla Q - S(\nabla u, Q) &= M_1 G,
\end{aligned}
\]  
(1.2)
where
\[
S(\nabla u, Q) = W \cdot Q - Q \cdot W + a(Q \cdot D + D \cdot Q) + \frac{2a}{3} \left( D - \frac{\nabla \cdot u}{3} \right) - 2a(D : Q) \left( Q + \frac{I}{3} \right),
\]
\[
\sigma(Q, G) = (Q \cdot G - G \cdot Q) - a(G \cdot Q + Q \cdot G) - \frac{2a}{3} G + 2a(Q : G) \left( Q + \frac{I}{3} \right),
\]
\[
G = -\frac{\delta E(Q)}{\delta Q} = K\nabla^2 Q - \left[ \alpha Q + \beta \left( Q^2 - \frac{\text{tr}(Q^2)}{3} I \right) \right] + \gamma \text{tr}(Q^2) Q,
\]  
(1.3)
and \( D = (\nabla u + \nabla u^T)/2 \) and \( W = \nabla u - \nabla u^T/2 \) are, respectively, the rate of strain and vorticity tensors. Moreover, the first two terms in \( S(\nabla u, Q) \) and the material derivative of \( Q \) define the Gordon-Schowalter derivative, \( G \) is the molecular field, \( \sigma(Q, G) \) the elastic stress tensor, and \( a \in [-1, 1] \) a geometric parameter of the nematic liquid crystal molecule — cf. [20]. We also impose the initial condition
\[
u(x, 0) = u_0(x), \quad Q(x, 0) = Q_0(x),
\]
and use one of the following boundary conditions:

1. \( u \) and \( Q \) are periodic on \( \partial \Omega \).
2. \( u|_{\partial \Omega} = 0, Q|_{\partial \Omega} = Q^0 \) or \( \partial_n Q|_{\partial \Omega} = 0 \).
This model is widely used in liquid crystal flows. The well-posedness of the Q-tensor model has been studied, yet there are no systematic efforts to analyse the variational and dissipative nature of the method. The governing system obeys the energy dissipation law and it is desirable to develop numerical schemes with the same property. Such numerical schemes are called the energy-stable schemes. The main challenge in this approach to the Q-tensor model is caused by the high nonlinearity of the bulk potential and the strong coupling between the velocity field and Q in the nonlinear stress term. 

Zhao and Wang employed a stabiliser in decoupled first-order and coupled second-order energy-stable schemes. Their main result is based on the assumption that the Hessian matrix of the bulk potential function is bounded from above. Additionally, in the decoupled first-order scheme, there is an artificially added term of order $\delta t$. Zhao et al. employed an "IEQ" strategy in a linear second-order coupled energy-stable scheme, but the proof of the main result has a gap. Here, a linear coupled unconditionally energy-stable scheme is constructed by following the "SAV" strategy from.

The rest of this paper is organised as follows. In Section 2, we derive a semi-discrete unconditionally energy-stable scheme and prove that the solution of the corresponding system is unique. Section 3 presents numerical examples to verify analytical results from Section 2. Our conclusions are in Section 4.

2. A Second-Order Linear Unconditionally Energy-Stable Scheme

Since the bulk energy $F_B(Q)$ is bounded from below, there is a constant $C_0$ such that

$$\mathcal{E}_1 = \int_{\Omega} F_B d\mathbf{x} + C_0 > 0.$$ 

Set $r = \sqrt{\mathcal{E}_1(Q)}$ and write the Eq. (1.3) in the form

$$G = \nabla \cdot (K \nabla Q) - \frac{r}{\sqrt{\mathcal{E}_1(Q)}} f_B(Q), \quad \text{(2.1)}$$

$$r_t = \frac{f_B(Q)}{2\sqrt{\mathcal{E}_1(Q)}} : Q_t, \quad \text{(2.2)}$$

$$f_B(Q) = \alpha Q + \beta \left( Q^2 - \frac{1}{3} \text{tr}(Q^2) \mathbf{I} \right) + \gamma \text{tr}(Q^2) Q. \quad \text{(2.3)}$$

The total energy can be rewritten as

$$\mathcal{E} = \int_{\Omega} \left( \frac{\mathbf{u}^2}{2} + \frac{K}{2} |\nabla Q|^2 \right) d\mathbf{x} + r^2 - C_0.$$ 

For the model with (2.1)–(2.3), we now consider a second-order scheme based on the backward differentiation formula with Adam-Bashforth explicit interpolation (BDF2).
More exactly, we set \( \tilde{f}^{n+1} = 2f^n - f^{n-1} \) and proceed as follows.

**Step 1.** Determine \((Q^{n+1}, r^{n+1}, \tilde{u}^{n+1})\) from the equations

\[
\begin{align*}
3Q^{n+1} - 4Q^n + Q^{n-1} &= 2\delta t (\hat{u}^{n+1} \cdot \nabla) \hat{Q}^{n+1} - S \left( \nabla \hat{u}^{n+1}, \hat{Q}^{n+1} \right) = M_1 G^{n+1}, \\
3\tilde{u}^{n+1} - 4u^n + u^{n-1} &= 2\delta t (\hat{u}^{n+1} \cdot \nabla) \tilde{u}^{n+1} + \nabla \cdot \sigma \left( \tilde{Q}^{n+1}, G^{n+1} \right) - \nabla \tilde{Q}^{n+1} : G^{n+1},
\end{align*}
\]

(2.4)

where

\[
G^{n+1} = \nabla \cdot (K \nabla Q^{n+1}) - \frac{f_b(Q^{n+1}}{\sqrt{\varepsilon_1(Q^{n+1})}},
\]

(2.5)

\[
3r^{n+1} - 4r^n + r^{n-1} = \int_\Omega 2 \sqrt{\varepsilon_1(Q^{n+1})} : (3Q^{n+1} - 4Q^n + Q^{n-1}) \, dx.
\]

(2.6)

**Step 2.** Find \((u^{n+1}, p^{n+1})\) from the equations

\[
\begin{align*}
3\frac{u^{n+1} - \tilde{u}^{n+1}}{2\delta t} + \nabla \left( p^{n+1} - p^n \right) &= 0, \\
\nabla \cdot u^{n+1} &= 0
\end{align*}
\]

(2.7)

(2.8)

(2.9)

with boundary condition \( u^{n+1} \cdot n |_{\partial \Omega} = 0 \).

Let us note the representation

\[
2(a^{n+1}, 3a^{n+1} - 4a^n + a^{n-1}) = |a^{n+1}|^2 + |2a^{n+1} - a^n|^2 + |a^{n+1} - 2a^n + a^{n-1}|^2 - |2a^n - a^{n-1}|^2
\]

(2.10)

used in the proof of the following theorem.

**Theorem 2.1.** The scheme (2.4)-(2.9) is unconditionally energy stable and satisfies the semi-discrete energy dissipation law

\[
\varepsilon_{DBF}^{n+1} - \varepsilon_{DBF}^n \leq -\delta t \left( M_1 \| G^{n+1} \|^2 + \eta \| \nabla \tilde{u}^{n+1} \|^2 \right),
\]

(2.11)

where

\[
\varepsilon_{DBF}^{n+1} = \frac{1}{2} \left( \frac{\| u^{n+1} \|^2}{2} + \frac{\| 2u^{n+1} - u^n \|^2}{2} \right) + K \left( \frac{\| \nabla Q^{n+1} \|^2}{2} + \frac{\| \nabla Q^{n+1} - \nabla Q^n \|^2}{2} \right)
\]

\[
+ \frac{1}{2} \left( \frac{|r^{n+1}|^2}{2} + \frac{|r^{n+1} - r^n|^2}{2} \right) + \frac{\delta t^2}{3} \| \nabla p^n \|^2.
\]

(2.12)

**Proof:** We consider the \( L^2 \)-inner products of the Eq. (2.4) with \( 2\delta t G^{n+1} \) and the (2.6) with \( 3Q^{n+1} - 4Q^n + Q^{n-1} \). Multiplying (2.7) and \( 2r^{n+1} \), and using the identity (2.10), we obtain

\[
(3Q^{n+1} - 4Q^n + Q^{n-1}, G^{n+1}) + 2\delta t ((\hat{u}^{n+1} \cdot \nabla) \tilde{Q}^{n+1}, G^{n+1})
\]
Consider now the A BDF2 Energy-Stable Scheme

\[ -2\delta t \left( S(\nabla \tilde{u}^{n+1}, \tilde{Q}^n), G^{n+1} \right) = 2\delta t M_1 \| G^{n+1} \|^2, \]

\[ (G^{n+1}, 3Q^{n+1} - 4Q^n + Q^{n-1}) \]

\[ = -r^{n+1} \int \frac{f_p(Q^{n+1})}{\Omega \sqrt{\varepsilon_1(Q^{n+1})}} : (3Q^{n+1} - 4Q^n + Q^{n-1}) \, dx \]

\[ - K \left( \frac{\| \nabla Q^{n+1} \|^2 + \| 2 \nabla Q^{n+1} - \nabla Q^n \|^2 + \| \nabla Q^{n+1} - 2 \nabla Q^n + \nabla Q^{-1} \|^2}{2} \right. \]

\[ \left. - \frac{\| \nabla Q^n \|^2 + \| 2 \nabla Q^n - \nabla Q^{-1} \|^2}{2} \right) \]

\[ \times 2 \left( \frac{|r^{n+1}|^2 + |2r^{n+1} - r^n|^2 + |r^{n+1} - 2r^n + r^{n-1}|^2}{2} - \frac{|r^n|^2 + |2r^n - r^{n-1}|^2}{2} \right) \]

\[ = r^{n+1} \int \frac{f_p(Q^{n+1})}{\Omega \sqrt{\varepsilon_1(Q^{n+1})}} : (3Q^{n+1} - 4Q^n + Q^{n-1}) \, dx. \]

In follows from the three equations above that

\[ K \left( \frac{\| \nabla Q^{n+1} \|^2 + \| 2 \nabla Q^{n+1} - \nabla Q^n \|^2 + \| \nabla Q^{n+1} - 2 \nabla Q^n + \nabla Q^{-1} \|^2}{2} \right. \]

\[ \left. - \frac{\| \nabla Q^n \|^2 + \| 2 \nabla Q^n - \nabla Q^{-1} \|^2}{2} \right) \]

\[ + 2 \left( \frac{|r^{n+1}|^2 + |2r^{n+1} - r^n|^2 + |r^{n+1} - 2r^n + r^{n-1}|^2}{2} - \frac{|r^n|^2 + |2r^n - r^{n-1}|^2}{2} \right) \]

\[ - 2\delta t \left( (\tilde{u}^{n+1}, \nabla) Q^{n+1}, G^{n+1} \right) + 2\delta t \left( S(\nabla \tilde{u}^{n+1}, \tilde{Q}^n), G^{n+1} \right) \]

\[ = -2\delta t M_1 \| G^{n+1} \|^2. \]  

(2.13)

Consider now the \( L^2 \)-inner products of (2.5) with \( 2\delta t \tilde{u}^{n+1} \) and (2.8) with \( u^{n+1} \), i.e.

\[ (3\tilde{u}^{n+1} - 4u^n + u^{n-1}, \tilde{u}^{n+1}) = -2\delta t \eta \| \nabla \tilde{u}^{n+1} \|^2 - 2\delta t (\nabla p^n, \tilde{u}^{n+1}) \]

\[ + 2\delta t (\nabla \cdot \sigma(Q^n, G^{n+1}), \tilde{u}^{n+1}) - 2\delta t (\nabla \tilde{Q}^n : G^{n+1}, \tilde{u}^{n+1}), \]  

(2.14)

\[ \frac{3}{4\delta t} (\| u^{n+1} \|^2 + \| u^{n+1} - \tilde{u}^{n+1} \|^2 - \| \tilde{u}^{n+1} \|^2) = 0. \]  

(2.15)

The Eq. (2.8) shows that for any divergence-free function \( v \) we have

\[ (u^{n+1}, v) = (\tilde{u}^{n+1}, v). \]  

(2.16)

The left-hand side of (2.14) can be written as

\[ (3\tilde{u}^{n+1} - 4u^n + u^{n-1}, \tilde{u}^{n+1}) \]

\[ = (3\tilde{u}^{n+1} - 3u^{n+1}, \tilde{u}^{n+1}) + (3u^{n+1} - 4u^n + u^{n-1}, \tilde{u}^{n+1}) \]

\[ = (3\tilde{u}^{n+1} - 3u^{n+1}, \tilde{u}^{n+1}) + (3u^{n+1} - 4u^n + u^{n-1}, u^{n+1}) \]
Considering the terms \( Q \) where 
\[
\text{The inequality (2.11) now follows from (2.12), (2.13) and (2.19).}
\]

Squaring this equation and multiplying the result by \( \delta \) yields 
\[
\frac{3}{4\delta t} (\|u^{n+1}\|^2 - \|\bar{u}^{n+1}\|^2) + \frac{\delta t}{3} (\|\nabla p^{n+1}\|^2 - \|\nabla p^n\|^2) = (\bar{u}^{n+1}, \nabla p^n).
\]

Combining the Eqs. (2.14)–(2.18), we obtain 
\[
\frac{\|u^{n+1}\|^2 + \|2u^{n+1} - u^n\|^2}{2} + \frac{\|u^n\|^2 + \|2u^n - u^{n-1}\|^2}{2}
\]
\[
\frac{\frac{2\delta t^2}{3} (\|\nabla p^{n+1}\|^2 - \|\nabla p^n\|^2) + \frac{\delta t}{3} \|u^{n+1} - \bar{u}^{n+1}\|^2}{2}
\]
\[-2\delta t \eta \|\nabla u^{n+1}\|^2 + 2\delta t \left( \nabla \cdot \sigma (\tilde{Q}^{n+1}, G^{n+1}), \bar{u}^{n+1} \right)
\]
\[-2\delta t \left( \nabla \tilde{Q}^{n+1} : G^{n+1}, \bar{u}^{n+1} \right).
\]

The inequality (2.11) now follows from (2.12), (2.13) and (2.19). \( \square \)

The Eqs. (2.4)–(2.7) show that the system in Step 1 is linear with respect to unknowns \( Q^{n+1} \) and \( \bar{u}^{n+1} \). We prove that the solution of this system is unique. Indeed, let us assume that for given \( Q^n, Q^{n-1}, u^n \) and \( u^{n-1} \) the system has two solutions \((Q_1, \bar{u}_1)\) and \((Q_2, \bar{u}_2)\). Considering the terms \( Q_0 = Q_1 - Q_2 \) and \( \bar{u}_0 = \bar{u}_1 - \bar{u}_2 \), we have 
\[
\frac{3Q_0}{2\delta t} + (\bar{u}_0 \cdot \nabla) \tilde{Q}^{n+1} - S (\nabla \bar{u}_0, \tilde{Q}^{n+1}) = M_1 G_0,
\]
\[
\frac{3\bar{u}_0}{2\delta t} + (\bar{u}^{n+1} \cdot \nabla) \bar{u}_0 = \eta \Delta \bar{u}_0 + \nabla \cdot \sigma (\tilde{Q}^{n+1}, G_0) - \nabla \tilde{Q}^{n+1} : G_0,
\]

where 
\[
G_0 = K \Delta Q_0 - \frac{f_2(\tilde{Q}^{n+1})}{\sqrt{\varepsilon_1(\tilde{Q}^{n+1})}} r_0,
\]
\[
r_0 = \int_\Omega \frac{f_2(\tilde{Q}^{n+1})}{2 \sqrt{\varepsilon_1(\tilde{Q}^{n+1})}} : Q_0 dx.
\]
Computing the $L^2$-inner products of (2.20) with $G_0$, (2.22) with $Q_0$ and (2.21) with $\tilde{u}_0$, and using the relationship (2.23), we obtain
\[
\frac{3}{2\delta_t}(Q_0, G_0) + (\tilde{u}_0 \cdot \nabla \tilde{Q}^{n+1}, G_0) - (S(\nabla \tilde{u}_0, \tilde{Q}^{n+1}), G_0) = M_1 \|G_0\|^2,
\]
and using the relationship (2.23), we obtain
\[
(G_0, Q_0) = -K\|\nabla Q_0\|^2 - 2|r_0|^2,
\]
\[
\frac{3}{2\delta_t}||\tilde{u}_0||^2 = -\eta\|\nabla \tilde{u}_0\|^2 + (\nabla \cdot \sigma(\tilde{Q}^{n+1}, G_0), \tilde{u}_0) - (\nabla \tilde{Q}^{n+1} : G_0, \tilde{u}_0).
\]
The three equations above yield
\[
\frac{3||\tilde{u}_0||^2}{2\delta_t} + 2|r_0|^2 + M_1 \|G_0\|^2 + K\|\nabla Q_0\|^2 + ||\tilde{u}_0||^2 = 0,
\]
so that $Q_0 = 0, \tilde{u}_0 = 0$.

Applying the Eq. (2.7), we eliminate the term $r^{n+1}$ in (2.13), thus obtaining
\[
\left(\frac{3}{2\delta_t} - M_1K\Delta\right)Q^{n+1} + \frac{M_1}{2}\hat{H}^{n+1} \int_\Omega \hat{H}^{n+1} : Q^{n+1} dx + \mathcal{L}_1(\tilde{u}^{n+1}) = g_1^n,
\]
where
\[
\mathcal{L}_1(\tilde{u}^{n+1}) = (\tilde{u}^{n+1}, \nabla) \tilde{Q}^{n+1} - S(\nabla \tilde{u}^{n+1}, \tilde{Q}^{n+1}),
\]
\[
\hat{H}^{n+1} = \frac{f_\delta(Q^{n+1})}{\sqrt{\varepsilon_1(Q^{n+1})}},
\]
\[
g_1^n = \frac{4Q^n - Q^{n-1}}{2\delta_t} - \frac{M_1(4r^n - r^{n-1})}{3} \hat{H}^{n+1} + \frac{M_1}{6} \hat{H}^{n+1} \int_\Omega \hat{H}^{n+1} : (4Q^n - Q^{n-1}) dx.
\]

Introducing the notation
\[
b^{n+1} = \int_\Omega \hat{H}^{n+1} : Q^{n+1} dx, \mathcal{L} = \left(\frac{3}{2\delta_t} - M_1K\Delta\right),
\]
we write the Eq. (2.24) as
\[
Q^{n+1} + \frac{M_1b^{n+1}}{2} \mathcal{L}^{-1}\hat{H}^{n+1} = \mathcal{L}^{-1}(g_1^n - \mathcal{L}_1(\tilde{u}^{n+1})).
\]
(2.25)

In order to determine the term $b^{n+1}$, we compute the inner product of (2.25) and $\hat{H}^{n+1}$. This leads to the representation
\[
b^{n+1} = \left(\int_\Omega \hat{H}^{n+1} : \mathcal{L}^{-1}(g_1^n - \mathcal{L}_1(\tilde{u}^{n+1})) dx\right) / \left(1 + M_1/2\right) \int_\Omega \hat{H}^{n+1} : \mathcal{L}^{-1}\hat{H}^{n+1} dx.
\]
(2.26)

Now we set
\[
\mathcal{L}_2(G^{n+1}) := \nabla Q^{n+1} : G^{n+1} - \nabla \cdot (\tilde{Q}^{n+1}, G^{n+1}),
\]
and write the Eq. (2.5) as

$$\frac{3\tilde{u}^{n+1}}{2\delta t} + (\tilde{u}^{n+1} \cdot \nabla) \tilde{u}^{n+1} - \eta \Delta \tilde{u}^{n+1} = g_2^n - \mathcal{L}_2(G^{n+1}).$$  

Step 1 in the above numerical scheme is equivalent to the Eqs. (2.25)–(2.27). In order to decouple $Q^{n+1}$ and $\tilde{u}^{n+1}$, we now use the Gauss-Seidel type iterations, so that

$$Q^{n+1,k+1} = \mathcal{L}^{-1}(g_1 - \mathcal{L}_1(\tilde{u}^{n+1,k})) - \frac{M_1 b^{n+1,k}}{2} \mathcal{L}^{-1}H^{n+1},$$

$$b^{n+1,k} = \int_{\Omega}^{H^{n+1}}: \mathcal{L}^{-1}(g_1^n - \mathcal{L}_1(\tilde{u}^{n+1,k})) \, dx,$$

$$\frac{3\tilde{u}^{n+1,k+1}}{2\delta t} + B(\tilde{u}^{n+1}, \tilde{u}^{n+1,k+1}) - \eta \Delta \tilde{u}^{n+1,k+1} = g_2^n - \mathcal{L}_2(G^{n+1,k+1}).$$

3. Numerical Results

In numerical simulations below, we assume that the directors of the LCPs are imposed in the $x$-$y$ plane, so that the tensor $Q$ is a $2 \times 2$ matrix. The unknown functions depend on variables $x$ and $y$ only and the model parameters are

$$a = 1.0, \quad \eta = 1, \quad \alpha = -0.2, \quad \gamma = 1, \quad K = 0.001, \quad M_1 = 1, \quad C_0 = 10.$$

3.1. Accuracy test

In the first example, we choose the initial and boundary conditions

$$u_0 = 0, \quad Q_0 = n_0 n_0^T - \frac{||n_0||^2}{2},$$

$$u|_{\partial\Omega} = 0, \quad Q|_{\partial\Omega} = Q_0|_{\partial\Omega},$$

where $n_0 = (\sin(2\pi x) \sin(2\pi y), 0)^T$. Additional data $u_1, Q_1$ can be obtained by the first-order scheme with a small time step. Avoiding the spatial discrete error, we employ 512 × 512 mesh grid on $[0, 1]^2$. The convergence order shown in Tables 1 and 2, approaches 2 if time step $\delta t$ tends to 0, consistent with theoretical analysis.

In the second example, we choose the initial and boundary conditions

$$u_0 = 0, \quad Q_0 = n_0 n_0^T - \frac{1}{2},$$

$$u|_{\partial\Omega} = 0, \quad Q|_{\partial\Omega} = Q_0|_{\partial\Omega},$$

where $n_0 = (\cos 2\pi x \sin 2\pi y)^T$ and consider the same computational domain and mesh size.


### 3.2. Development of the defects

The next examples deal with the defect development. We start with the defect at the point +1 subject to the Dirichlet boundary condition on $Q$. For the domain $[0, L_x] \times [0, L_y]$, $L_x = L_y = 2$, we consider a no-slip boundary condition for $u$, $u|_{\partial \Omega} = 0$ and a Dirichlet boundary condition for $Q$, viz.

$$Q|_{\partial \Omega} = \frac{\hat{n}_0 \hat{n}_0^T}{\|\hat{n}_0\|^2} - \frac{I}{2}, \quad \hat{n}_0 = (x - 0.5L_x, y - 0.5L_y)^T$$

and the initial value

$$Q_0 = \frac{n_0 n_0^T}{\|n_0\|^2} - \frac{I}{2}, \quad n_0 = (x - 0.25L_x, y - 0.25L_y)^T.$$ 

Fig. 2(a) shows an initial +1 type defect at the point (0.25, 0.25). The boundary condition breaks the initial defect into two +1/2 type defects moving away from each other and reaching a stable state. In Figs. 2(b)-2(e), the line segments indicate the director of the LC and the colors show the parameter orders. The order parameters at the defects are almost zero — cf. blue points in Figs. 2(b)-2(e). Fig. 1 shows that the system energy decreases with time, consistent with our analysis. The velocity at $t = 50, 100, 200$ is shown in Figs. 2(f)-2(h).

In this example, we use the initial data

$$Q_0 = \frac{n_0 n_0^T}{\|n_0\|^2} - \frac{I}{2}, \quad n_0 = \begin{cases} (1, 0)^T, & \sqrt{(x - 1)^2 + (y - 1)^2} < 0.4, \\ (0, 1)^T, & \text{otherwise}. \end{cases}$$
Figure 1: Energy versus time.

Figure 2: (a) shows the director of initial data $Q_0$; (b)-(e) show the director and order parameters (in color) for $t = 50, 100, 150, 200$, respectively; (f)-(h) show the velocity at $t = 50, 100, 200$, respectively.
and the boundary conditions

\[ Q|_{\partial \Omega} = \frac{\tilde{n}_0 \tilde{n}_0^T}{\|\tilde{n}_0\|^2} - \frac{1}{2}, \quad \tilde{n}_0 = (0, 1)^T, \quad u|_{\partial \Omega} = 0. \]

We also consider the liquid crystal dynamics in situations with and without hydrodynamic effects for the same set of parameters and present the corresponding results in Figs. 3 and 4, respectively.

There is a mismatch in orientation in the circular region initially. In Fig. 3, we use the line segment to indicate the directors and the colors to indicate the order parameters. The graphs show that the order parameters are close to 0 near the defect points (the blue points). The mismatch produces two \( +1/2 \) type defects (left and right) and two \( -1/2 \) type defects (top and bottom). Defects approach each other and disappear. The directors become homogeneous perpendicular to the \( y \)-boundary.

Fig. 4 shows that the radius of mismatch ring reduces as time developed — cf. Figs. 4(a)-4(c)). Finally, the mismatch disappears — Fig. 4(d).

Fig. 5 demonstrates energy dissipation in both cases. It agrees with the analysis in Section 3. System with hydrodynamic effects admits easier development into homogeneous nematic state.

![Figure 3: Director and order parameters (in color). Figures (a), (b), (c) and (d) respectively correspond to \( t = 10, t = 25, t = 50 \) and \( t = 100 \).](image-url)
In the last example, we examine the impact of anchoring boundary conditions on the liquid crystal dynamics. The initial conditions are the same as before but the boundary conditions are changed. The computational domain is $[0, L_x] \times [0, L_y]$, $L_x = 2, L_y = 1$ and there is an initial defect of $+1$ type at the point $(1.0, 0.5)$. The forces on the $x$- and $y$-axes exerted on point $(1.0, 0.5)$ are different and the initial defect is unstable. The strong
anchoring boundary condition for $Q$ in situation considered in Fig. 6 has the form

$$Q|_{\partial \Omega} = \frac{\hat{n}_0 \hat{n}_0^T}{\|\hat{n}_0\|^2} - \frac{I}{2}, \quad \hat{n}_0 = (x - 0.5L_x, y - 0.5L_y)^T.$$  

The initial defect breaks into two $+1/2$ type defects. The $y$-axis boundary is closer to the defect than the $x$-axis and anchoring is strong. Therefore, the break is easier in the $x$-axis.

Figure 6: The director and order parameters under strong anchoring. $t = 5, 50, 100, 150$.

Figure 7: The director and order parameters under weak anchoring. $t = 5, 50, 100, 200$.
direction. In situation considered in Fig. 7, the weak anchoring boundary condition for $Q$ has the form

$$\frac{\partial Q}{\partial n} = 0.$$  

This defect also breaks up into two $+1/2$ type defects. However, because of the weak anchoring on the boundary, the defect breaks easier in the $y$-axis direction. Fig. 8 shows the energy dissipation of the systems under weak and strong anchoring effects. The results are consistent with the analysis of Section 3.

![Energy dissipation graphs](image)

Figure 8: Energy versus time for different anchoring conditions.

4. Conclusion

We construct a BDF2 linear energy-stable numerical scheme for a general tensor-based model of liquid crystals and test it on several examples. Numerical results show the second-order convergence and energy dissipations, consistent with the theoretical analysis. We also compare the dynamics of the defects with different hydrodynamic effects and anchoring conditions.

Acknowledgments

This work is partially supported by the National Natural Science Foundation of China (Grant Nos. 11571045, 11671052, 11871105), by the State Scholarship Fund of CSC (Grant No. 201406045016) and by the Fundamental Research Funds for Central Universities (Grant No. 2015KJJCB23).

References


